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STRENGTHENING OF ROCK AGAINST SHOCK EFFECTS DASA SUB-TASK NO. 13.192 TESTS FOR STRENGTH CHARACTERISTICS OF ROCK PILEDRIVER PROJECT



MRD LABORATORY NO. 64/90 SEPTEMBER 1964

CORPS OF ENGINEERS
U.S.ARMY

MISSOURI RIVER DIVISION LABORATORY

OMAHA, NEBRASKA



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PREFACE

The studies reported herein were authorized by the Chief of Engineers (ENGNC-EM) and were initially funded in FY 1964 by sub-allotment from the U. S. Army Corps of Engineers Waterways Experiment Station through the U. S. Army Corps of Engineers Ohio River Division Laboratories for research in techniques of rock strengthening.

Additional funds were received in FY 1964 from the Defense Atomic Support Agency, DASA MIPR 571-64, by sub-allotment from the U. S. Army strengthening for the PILEDRIVER project under DASA NWER Sub-Task No. 13.192 "Strengthening of Rock Against Shock Effects."

The initial emphasis of the studies conducted and reported herein are for the purpose of establishing the strength characteristics of intact and jointed rock necessary for the design, installation, and evaluation of rock strengthening systems to be incorporated in the PILEDRIVER project.

The laboratory tests were performed at the U. S. Army Corps of Engineers Missouri River Division Laboratory by members of the Physical Tests and Investigations Section under the supervision of Messrs. E. J. Deklotz and Wm. J. Heck, who also jointly prepared this report. Mr. L. A. Brown was Director of the Missouri River Division Laboratory during the conduct of the program.

Additional studies to develop strength characteristics of the fractured (jointed) rock bonded with chemical adhesives and to develop a preliminary rock bolting theory have been conducted by the U. S. Army Corps of Engineers Ohio River Division Laboratories and U. S. Army Corps of Engineers Omaha District respectively as a part of the DASA NWER Sub-Task No. 13.192 and will be presented in separate reports by those agencies.

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SUMMARY

The strength characteristics of the quartz monzonite has been determined by direct tensile, unconfined compression, and triaxial compression tests. The data indicate that the rock behaves elastically and retains its brittleness throughout the range of confining pressures used. In general, the rock is uniform in composition and structure, having an average bulk, dry density of 2.66 and an average porosity of 0.27 percent. Feldspar and quartz are the principal mineral constituents with a small amount of biotite making up most of the remainder of the rock.

The Mohr strength envelop for the intact rock and rock with healed joints that acted as intact rock, determined by using the average tensile strength of 1,450 psi, an average unconfined compressive strength of 30,530 psi, and average maximum principal stresses of 32,690, 36,340 and 46,050 psi at 150, 450, and 1,350 psi confining pressures respectively, is a slightly curved line from which a coefficient of internal friction angle of 56 degrees and a cohesive strength of 3,600 psi is obtained.

The angle of friction was found to be 31 degrees for the natural open joints and 28 degrees for sawed joints as determined from a plot of the points of shear and normal stresses on these pre-established planes. These values indicate that small irregularities on the joint surface have little effect on the coefficient of friction. All of the open and sawed joint surfaces showed the development of slickensides and granulation of mineral grains.

Statistical evaluation of the limited number of test results indicates that in general the rock has rather uniform strength characteristics and that the test data are reasonably consistent. The test results all fall within 95 percent Confidence Limits, assuming <u>t</u> distribution and using Student's <u>t</u> values. Coefficient of variation ranged from 3.3 percent for unconfined compression tests to 28.2 percent for the tensile strength tests. Values of 10.8 to 16.8 percent were obtained for the intact rock triaxial compression tests.

TESTS FOR STRENGTH CHARACTERISTICS OF ROCK

PILEDRIVER PROJECT

INTRODUCTION

l. This report presents results of tests for shear, compressive, and tensile strength characteristics of rock of the type which will be encountered and subject to rock strengthening techniques and evaluation as part of Piledriver Project. Triaxial compression tests were made of intact rock and of rock having natural healed or open joints, and joints smoothly sawed in the laboratory. Unconfined compression and direct tension tests were made of intact rock and of rock containing natural healed joints. Certain testing and instrumentation techniques were developed and are described in detail where it is considered the information may be of value in future work of a similar nature.

ROCK IDENTIFICATION

2. Sections of the 2 3/8 inch diameter rock core, from Exploratory Drill Hole U1501-U1 at the project site were submitted for test. Top elevation of hole is 4254.7 feet. The hole is an inclined boring at a downward angle of 210 from horizontal. Throughout this report all depths are inclined depths of core along the boring. Only that portion of rock core which lies in the quartz monzonite was submitted for test. The quartz monzonite is generally a light gray, dense, porphyritic rock with a fine to medium-grained groundmass. Orthoclase and plagioclase feldspars with quartz are the most abundant minerals; however, a small but varying amount of biotite mica occurs as scattered flakes or clusters. Occasionally large, pink crystals of feldspar up to 2 1/2-inches in length are present. Very little alteration and weathering was noted in the core specimens tested. Most of the healed joints observed in the rock are well cemented with quartz and feldspar along with a small amount of calcite. Pyrite was also a common mineral in some of the healed joints. A few of the cores were partially healed with a water soluble mineral which dissolved when the cores were soaked prior to test; consequently these cores separated at the joint and had to be tested as open joint rock specimens. The natural open joints were generally free of mineral deposits or rock dust and their surfaces, although rough, were essentially planar.

TEST PROGRAM

3. Core specimens were selected for test as follows:

Triaxial Compression Test Specimens

1

1,700

Condition of Rock Core	Core No.	Confining Angle of Joint from Pressure, Psi Horizontal, decrees
Intact Rock	37, 133, 144 40, 46, 69 28, 83, 136	150 - 450 - 1350 -
Natural Healed		
Joints	130, 105, 142, 143 131, 151, 138, 159 155, 134, 153	150 41, 55, 60, 62 & 69 450 44, 55, 61, 64 1350 51 , 57, 64
Natural Open		•
Joints	120, 127, 158, 108, 12 169, 107, 164 112 159	5 150 38, 54, 59, 62, 66 450 44, 62, 66 1350 50 3000 64
Sawed Joints	23 22, 11, 12, 18, 27 128, 129, 35	150 55 450 45, 62 1350 45, 55, 62 3000 45, 55, 62
Unconfined Comp	pression Test Specimens	
Intact Rock	39, 86, 132	
Natural Healed Joints	150, 137	- 56, 63
Direct Tension	Test Specimens	
Intact Rock	9, 88, 126	•
Natural Healed Joints	30, 43, 145, 54	- 54, 65 & 90, 66 & 90, 90
Cyclic Loading	in Unconfined Compressi	<u>00</u>
Intact Rock	47	•

Pieces of core cut from the ends of the several strength test specimens identified above were tested for bulk saturated surface-dry specific gravity, bulk dry density, and porosity.

4. The stress-strain data presented graphically in this report can be made available in tabulated form upon request.

PROCEDURE AND RESULTS

Specific Gravity, Density, and Porosity

Procedure

1

- 5. In the preparation of the strength test specimens, the longest end piece cut from each strength test core was squared on both ends and used in the determination of bulk saturated surfacedry specific gravity, measured bulk dry density, and porosity. Lengths of these end pieces varied from 1.16 to 4.35 inches. According to the procedure provided in the authorizing test requests, each specimen was measured for length at quarter points and for diameter at third points about the circumference and the volume computed from these measurements. The specimens were permitted to absorb water at room temperature by immersion within a pressure vessel for 1/2-hour at 1200 psi pressure and, upon completion of the saturation period, were individually weighed in water, surface dried, and again weighed in air. The oven dry weight was determined after drying for 72 hours at 230 F. The several parameters were computed as follows:
 - a. Specific gravity, bulk saturated surface-dry

$$g = \frac{B}{B-D}$$

where -

g = Specific gravity, bulk saturated surface-dry

B = Weight in grams of saturated surface-dry specimen
 in air

D = Weight in grams of saturated specimen in water (saturated by immersion in water for 1/2-hour under 1200 psi pressure)

b. Measured bulk dry density

$$d = \frac{A}{V_1}$$

where -

d = Measured bulk dry density

A = Weight in grams of oven dried specimen in air

c. Porosity

$$p = \frac{V_2}{V_3} \times 100$$

where -

 V_2 = Void volume, B-A

 $V_3 = Solid volume, V_1 - V_2$

Results

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6. Results of these tests are presented in Table 1, and indicate that the quartz monzonite is very uniform in composition and structure.

Triaxial Compression Tests

Procedure

- 7. Tests in triaxial compression were made to determine the shear strength of the several conditions of rock. All cores were cut to an L/D ratio of 2, ends ground and lapped, and immersed for 72 hours in water at room temperature before testing. Particular care was taken in cutting and finishing the ends of the cores to assure that they were approximately normal to the axis and that essentially parallel, flat, smooth end surfaces were obtained. Final finishing was accomplished by hand lapping on a glass plate with No. 600 carborundum compound. The sawed joints were produced with a diamond saw at the preselected angles and rough ground with No. 240 carborundum to remove the saw grooves.
- 8. The triaxial compression testing was conducted with the apparatus shown in Plate 1 in which confining pressures up to 10,000 psi can be maintained while axial loads up to 400,000 pounds are applied with a universal hydraulic testing machine. Axial strain was measured by a 1/10,000-inch dial gage. Corrections were made for the deformation of the hardened steel parts (Rockwell C 58 or harder), in order to determine the true strain of the rock specimen during test. All of the core specimens were surface dried after the 72-hour immersion period. A thin (15 mil) sheet of vinyl plastic was placed around the core and the top and bottom steel platens to prevent punctures at the interface. The assembly was then covered with a thick rubber membrane made from a section of motor bike inner tube and sealed with rubber 0 rings to prevent oil from entering the specimen.

9. After the triaxial apparatus was completely assembled, anti-foaming hydraulic oil was pumped into the chamber until it drained from the overflow tube and air bubbles ceased to rise. The confining pressure was raised to the desired value and maintained constant throughout the test by manual operation of the screw piston device attached to the base of the apparatus. Axial loads were applied to produce, as nearly as possible, a rate of strain of 1/1000 inch per minute in the specimen until failure occurred. Preestablished joint or failure planes were subjected to multi-stage triaxial tests in which, after failure, the confining pressure was raised to the next higher preselected value and the test continued. In some cases this procedure was repeated a number of times.

Results

- 10. Stress-strain curves and companion Mohr diagrams for the several test conditions are shown in Figures 1 through 78 with a brief description of each core specimen and a diagram of the manner of failure. Photographs of cores showing typical failure planes after test are presented in Plates 2 and 3.
- ll. The modulus of elasticity, E, is determined as the slope of the straight line portion of the stress-strain curve and the value is shown with each curve. The average values of E of intact rock cores and of cores with healed joints which failed as intact rock are:

Confining pressure, psi, 150 450 1350 Modulus of elasticity, 10⁶ psi, 11.75 11.74 11.38

- 12. The E values for the unconfined compression and tensile test specimens determined by strain gage measurements are 10.46×10^6 and 8.48×10^6 psi, respectively.
- 13. A statistical summary of the triaxial compression test results for the average maximum principal stresses of five specimens, three of intact rock and two of well-healed jointed rock, at each of the selected confining pressures is as follows:

Confining pressure, psi, Maximum principal stress, psi,	150 32,690	<u>450</u> 36,340	1.350 46.050
Standard deviation, psi,	5,210	3,910	7,750
Coefficient of variation, percent,	15.9	1C.8	16.8
95 percent Confidence Limits,	19,240 to	25,490 to	24,540 to
Student's t distribution, psi, *	47,140	47,190	67,560

^{*} U. S. Army Engineer Waterways Experiment Station, CE, <u>Basic</u>
<u>Statistical Definitions and Procedures</u>. Miscellaneous Report No. 2-250 (Vicksburg, Miss., January 1958).

Unconfined Compression Tests

Procedure

14. Specimens for unconfined compressive strength tests were cut, prepared, and immersed in water prior to test in the same manner as the triaxial compression test specimens described in paragraph 7 above. Axial and diametric strains were measured by use of Type A-1 SR-4 strain gages cemented to the specimens with Eastman 910 cement. Three axial gages were equally spaced around the specimen at the center of its height, and three diametric gages were similarly located around the diameter. Specimens were loaded through a high strength steel spherical head and solid pedestal of Rockwell Hardness C 58, at a rate of 50 psi/sec. The bearing face of the spherical head and solid pedestal is 2.42 inches in diameter and is ground flat to 0.0005-in. No capping material was used on the test specimens. Plate 4 is a photograph of one of the test specimens in the testing machine immediately prior to loading.

Results

15. Results of these tests are graphically presented in Figures 79 through 83. Poisson's Ratio was computed using values taken from the curves for diametric and axial strain. Ignoring infinitesimals of the second order it can be shown that the volumetric strain is equal to the axial strain plus 2 times the diametric strain. The stress-volumetric strain curve appears to be of interest as it seems to provide a means for determining the "yield point" of the rock which can be defined as that point on the curve at which the slope of the tangent changes sign. The curve seems to indicate further a consolidation of particles as the load increases up to this point; beyond this point the rock structure begins to disrupt and tear apart. For the 5 specimens tested using strain gage instrumentation, it will be observed that this point ranges between approximately 20,000 and 25,000 psi. Two additional specimens, Core Nos. 32 and 84, were tested for compressive strength with no strain gage instrumentation. These cores showed ultimate strengths of 29,480 and 32,200 psi, respectively. The results of all seven tests are summarized as follows:

Maximum Compressive Strength, Core No. 84, psi,	32,200
Minimum Compressive Strength, Core No. 32, psi,	28,770
Average Compressive Strength, psi,	30,530
Standard Deviation, psi,	1,162
Coefficient of Variation, percent,	3.3
95 percent Confidence Limits, Student's t distribution, psi,	27,690 to 33,370

Direct Tension Tests

Procedure

16. All direct tension test specimens were immersed in water for 72 hours at room temperature prior to test. Generally the test specimens were cut to an L/D ratio of 2 and cemented into metal end caps using an epoxy resin material identified as Plastic Steel, Devcon A - Putty Type, produced by the Devcon Corporation, Danvers, Mass. After allowing to harden for 24 hours, the assembly, with SR-4 strain gages attached in the same configuration as for the unconfined compression test specimens, is suspended by means of heavy duty roller chain in the testing machine and load applied to failure. Plate 5 is a photograph of a specimen immediately prior to test. The upper and lower roller chain are at right angles to each other in order to reduce eccentricity and bending in the specimen.

Results

17. Stress-strain curves are shown in Figures 84 through 90. Initially the tests were conducted using full cross-section specimens; however, with the development of failures in the cap (Core No. 54, Figure 90) which were considered to be a manifestation of end restraint, it was decided to test reduced section specimens. The reduced sections were cut by means of a carbide tipped tool in a machine lathe operating at slow speed with a minimum amount of water. Improvement was shown in type of break by this method but, under this condition, one end break (Core No. 126, Figure 86) did occur. One specimen, Core No. 51, was tested for tensile strength with no strain gage instrumentation and showed an ultimate tensile strength of 2,353 psi. The results of all eight tests are summarized as follows:

Maximum Tensile Strength, Core No. 51, psi,	2,353
Minimum Tensile Strength, Core No. 145, psi,	1,026
Average Tensile Strength, psi,	1,452
Standard Deviation, psi,	410
Coefficient of Variation, percent,	28.2
95 percent Confidence Limits, Student's t distribution, psi,	482 to
	2,422

18. In view of the values obtained for compressive strength, the difficulties encountered in developing a truly direct tensile load on the specimens, and the strength envelops to be discussed later, it is considered that the strength shown by Core No. 51 may most nearly represent the true tensile strength of the rock. The strength of the remaining 7 cores all appear rather low.

Cyclic Loading in Unconfined Compression

Procedure

19. One specimen, Core No. 47, was cyclic loaded to failure in unconfined compression. The specimen was prepared for test in the same manner as described for the conventional unconfined compression test specimens. The first three loadings were to nominally 20,000 psi. Subsequent duplicate loadings were increased in approximately 2,000 psi increments beyond the original 20,000 psi loadings and failure occurred at the beginning of load release during the eleventh cycle. Strain readings were taken upon load release as well as during load application.

Results

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20. Stress-strain data are tabulated in Tables 2 and 3. Figures 91 and 92 are stress-strain curves for a representative selection of the several loadings. It is not entirely clear why the axial strain values show a strain reversal (tension) upon load release unless perhaps this is a manifestation of strain gage slippage. If this is true, a change to epoxy resin type cement might correct this condition. Further investigation in this area seems to be indicated.

GENERAL DISCUSSION OF TEST RESULTS

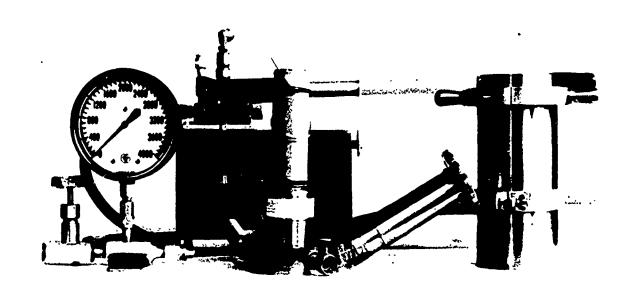
- 21. When the shear strengths on a pre-established plane such as an open or sawed joint are plotted as in Figure 93, a strength envelop of best fit of the points is obtained from which the coefficient of friction angle is determined. It is interesting to note that the angle of friction of sawed and natural joints are very similar (28 to 31 degrees), being only 3 degrees apart. The points plotted represent points of failure at all stages of confining pressure. Generally, there appears to be little difference between the angles of friction obtained from plots of the first stage and subsequent stages. Furthermore the coefficient of friction is fairly constant for the range of joint angles tested. This would indicate that the coefficient of friction is reasonably constant for this rock and unless the joint plane surfaces are extremely nonplanar small surface irregularities have very little effect in increasing the angle of friction. Small slickensides and granulation of mineral grains were developed on the joint plane surfaces.
- 22. The Mohr envelop for the intact rock was determined from the average strengths of a group of 5 cores for each confining pressure and includes the averages of those tested in tension and unconfined compression. Since the rock cores with well-healed joints behaved as intact rock when tested in triaxial compression,

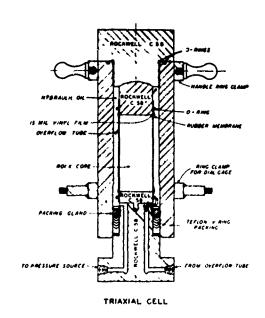
their results have been included with this group. Joints healed with quartz and feldspar appeared to be as strong as the intact rock, while those filled primarily with pyrite are somewhat weaker. For the three confining pressures used the strength envelop shows a coefficient of internal friction angle of about 56 degrees and a cohesive strength of about 3,600 psi. The envelop tends to be practically a straight line but becomes slightly curved when fitted to the unconfined compression and tensile strength values. A straight line projection in this area of the curve indicates a tensile strength of around 2,500 psi. This more nearly checks the value of 2,353 psi shown by Core No. 51.

23. Generally the observed shear angles approach those determined from the formula $45^{\circ} + 2$ (73 degrees) with only a few degrees difference. In most cases, the measured shear plane angles of the unconfined compression tests show the greatest variation, about 7 degrees less than the computed values. It was only with the confining pressure of 1,350 psi that the core specimens tended to shear along a single plane without vertical tension fractures. For the most part, the test results appear to fit Mohr's criterion quite well.

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TRIAXIAL TEST APPARATUS

PLATE 1

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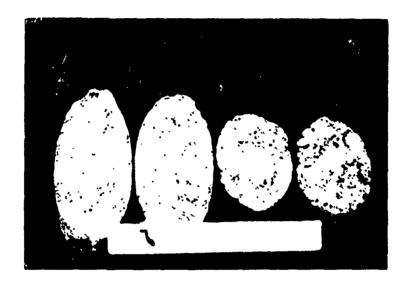
Intact Rock



Natural Healed Joints

TYPICAL ROCK CORES AFTER TEST

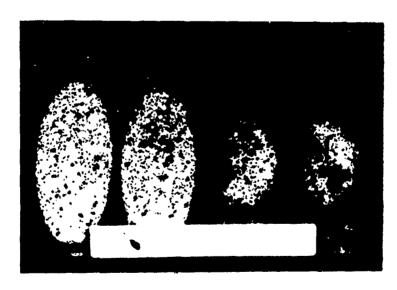
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Natural Open Joints



Sawed Joints

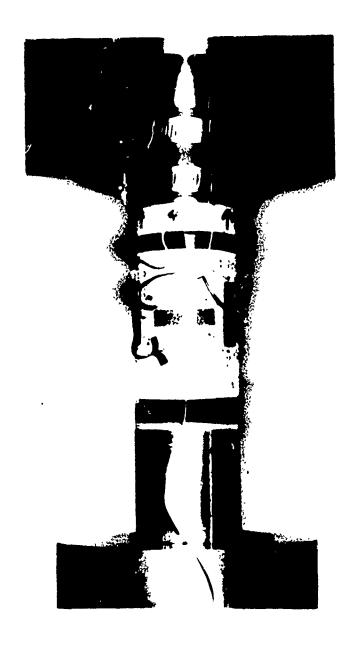
TYPICAL ROCK CORES AFTER TEST

PLATE 3

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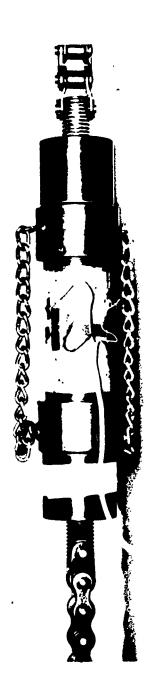
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UNCONFINED COMPRESSION TEST ASSEMBLY

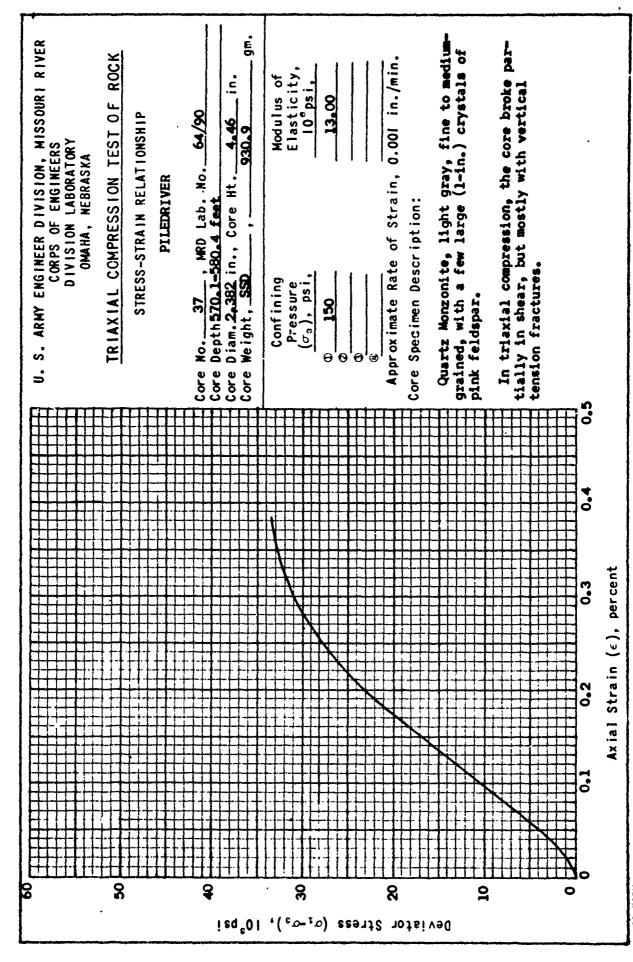
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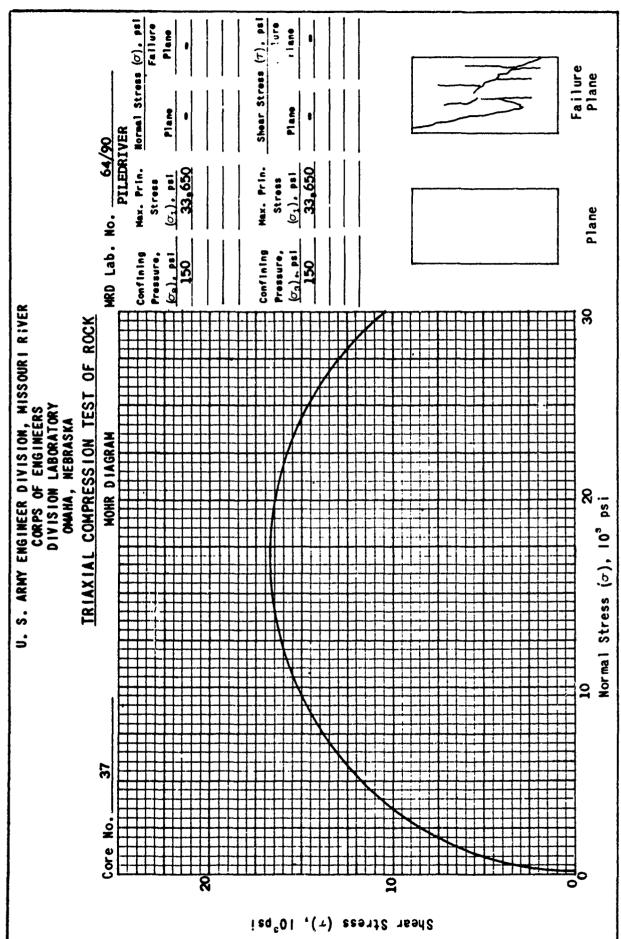
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DIRECT TENSION TEST ASSEMBLY

PLATE 5





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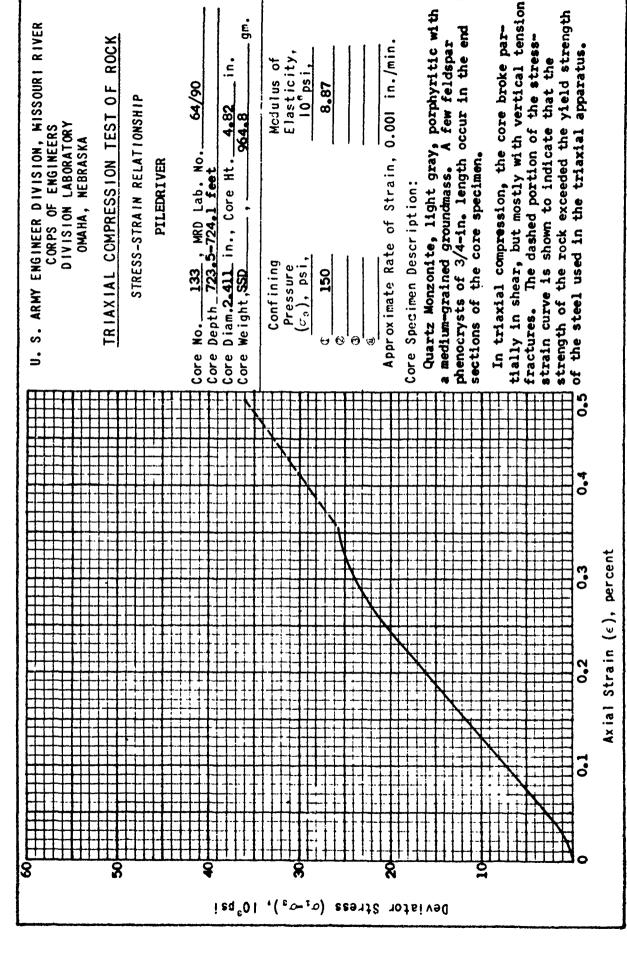
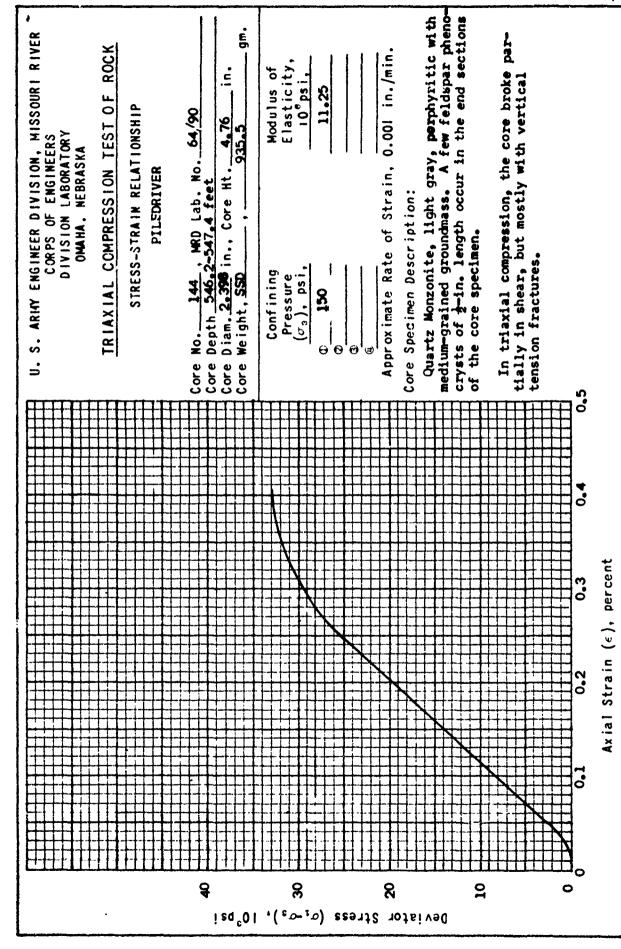


FIGURE 4

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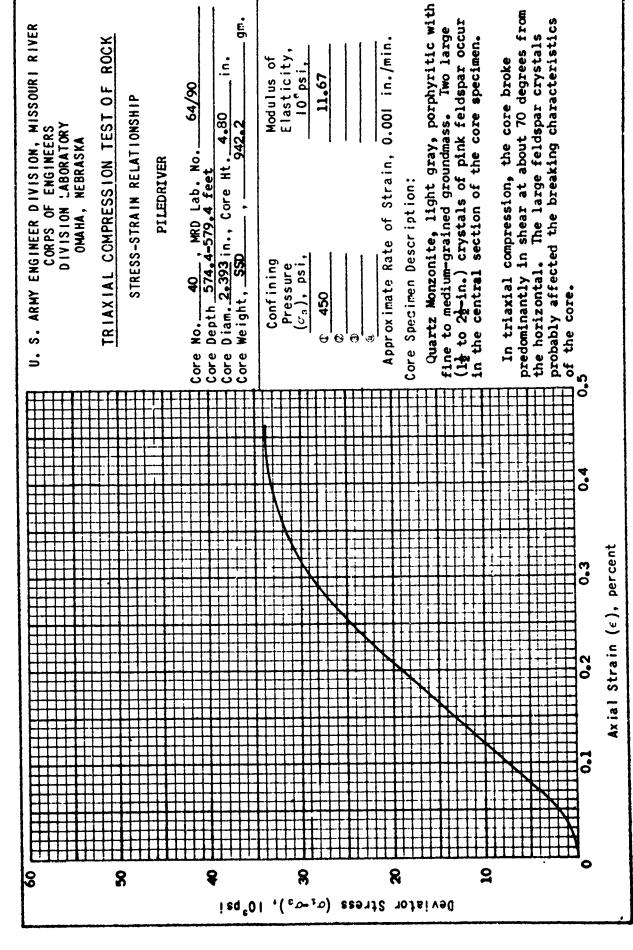
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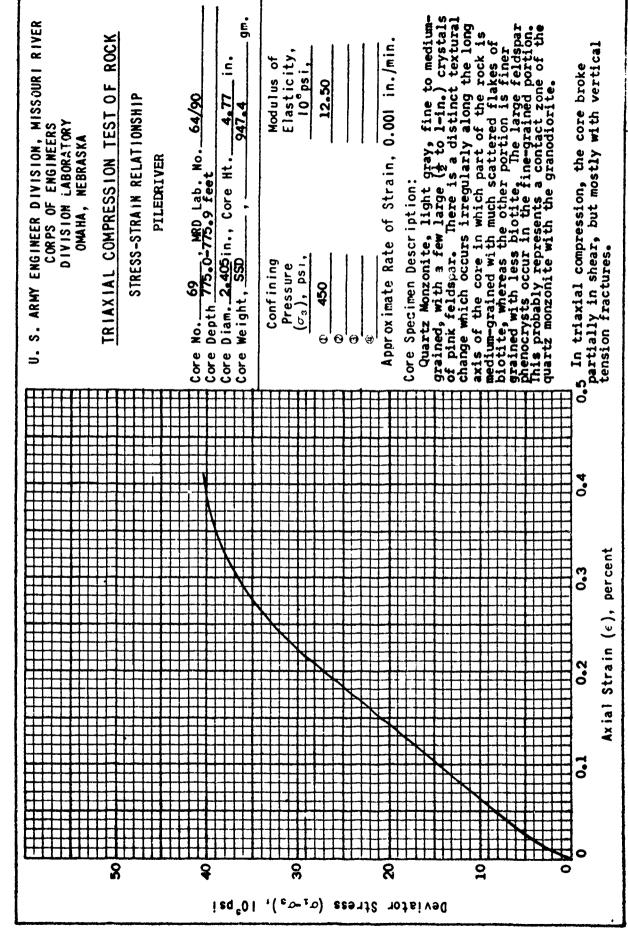
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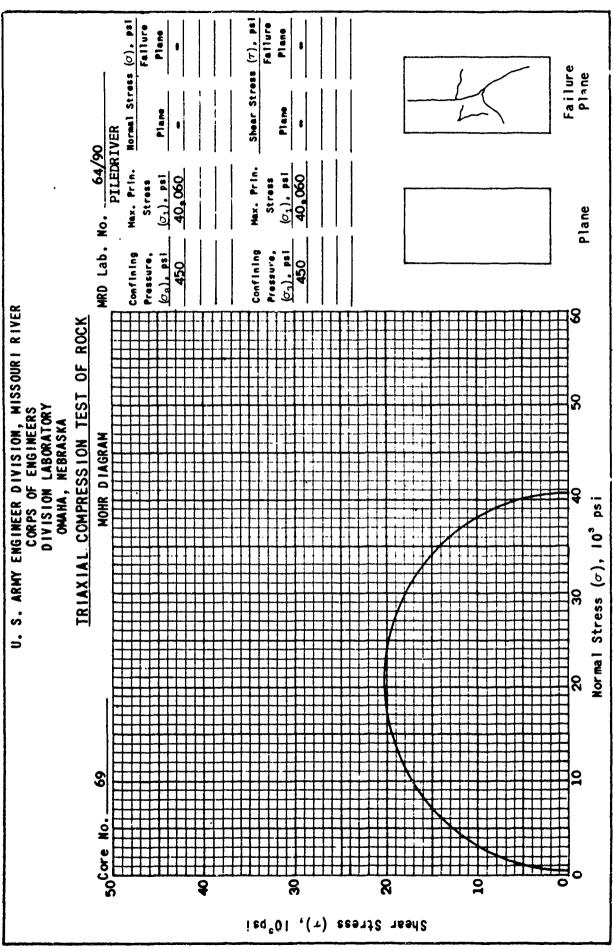
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Deviator Stress $(\sigma_z - \sigma_s)$, $10^s psi$

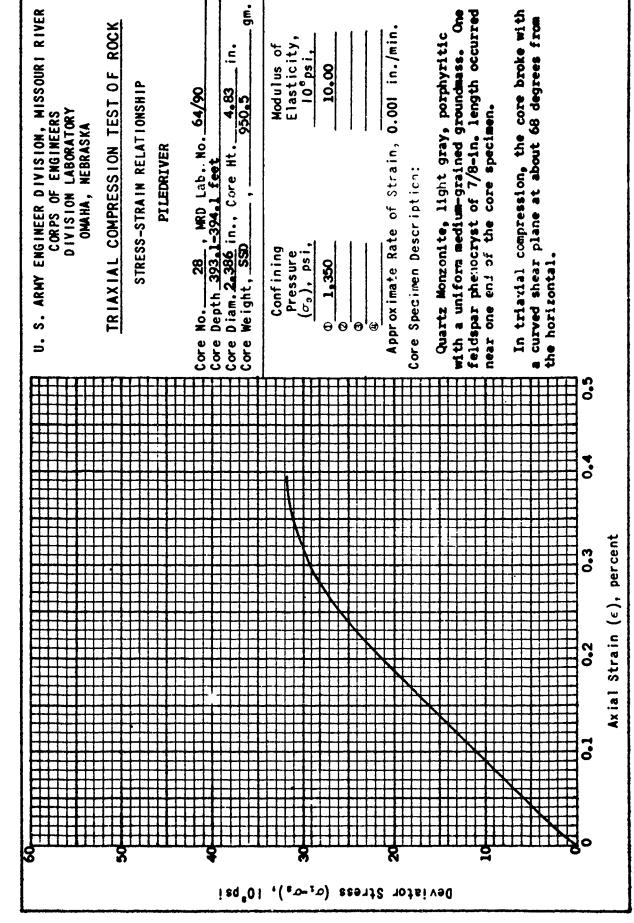
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FIGURE 10

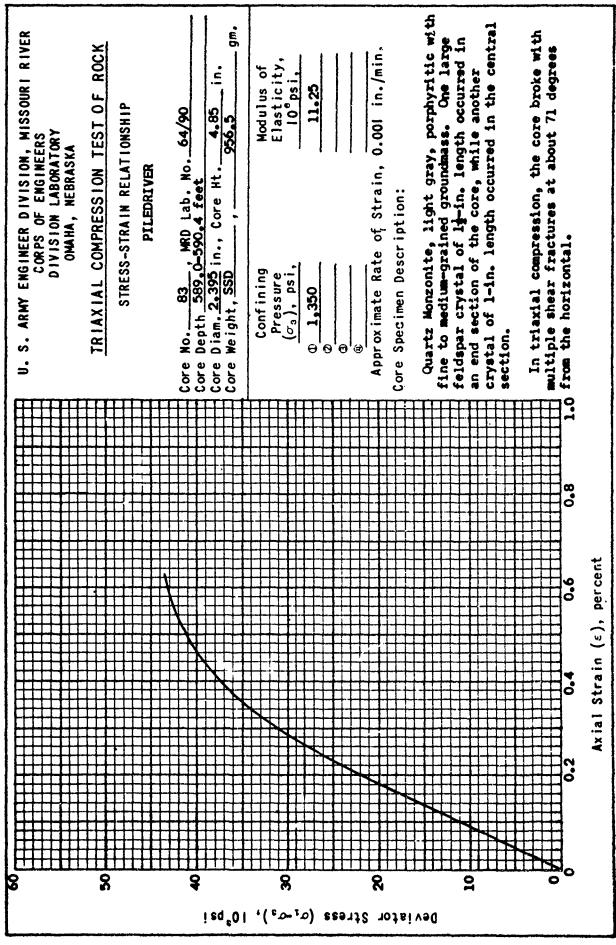




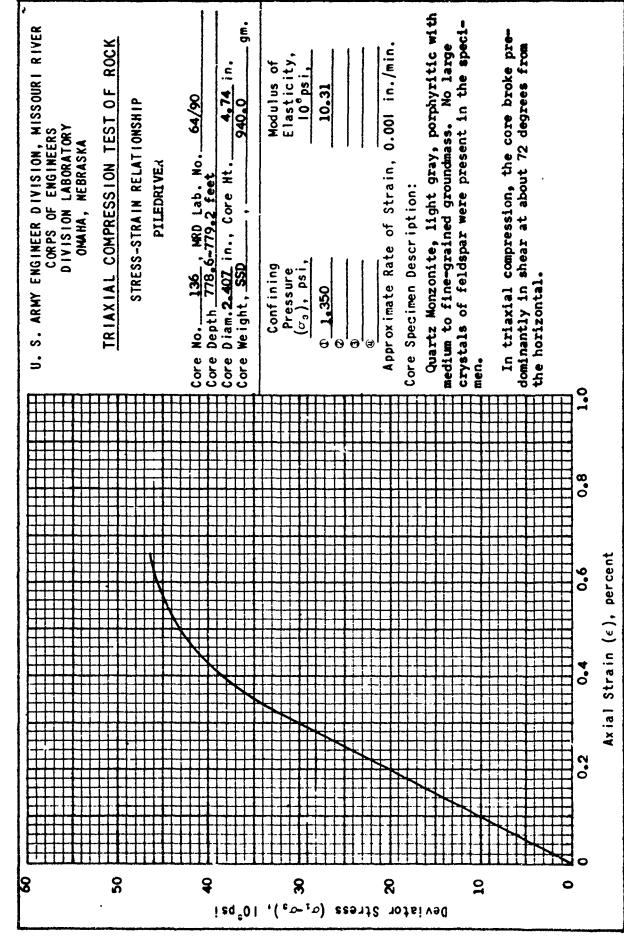
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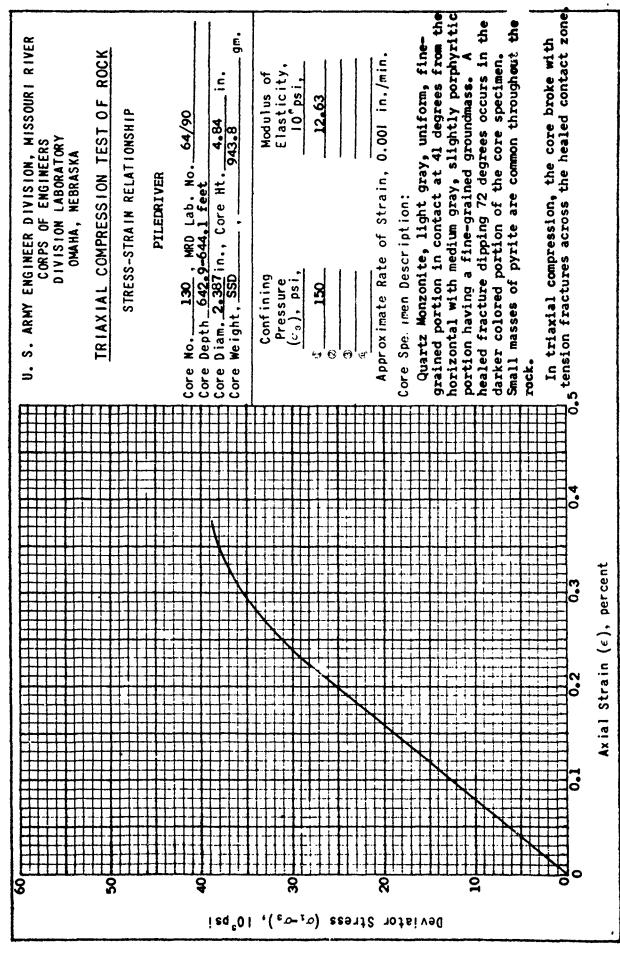
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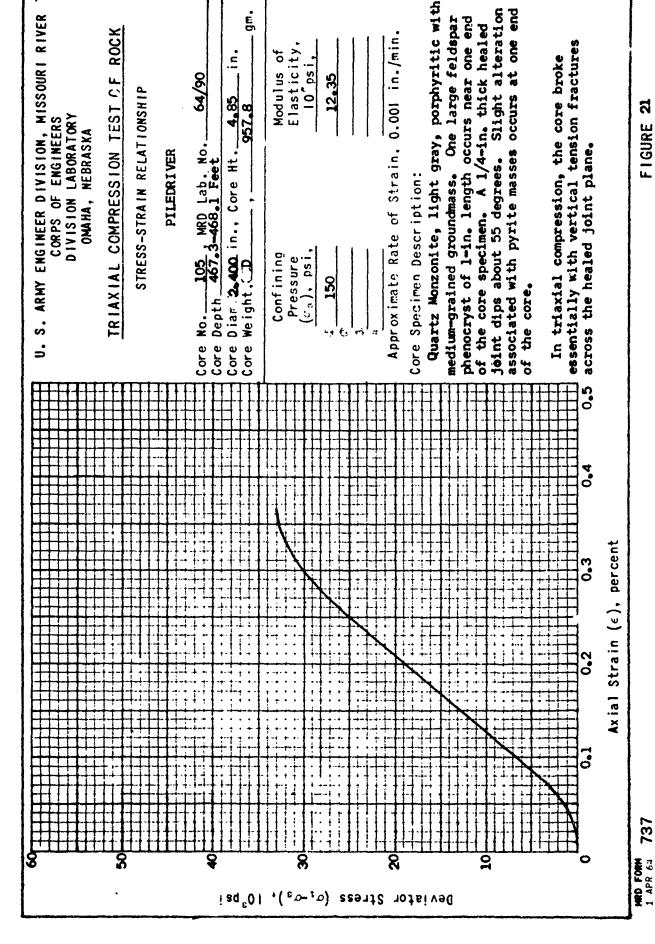


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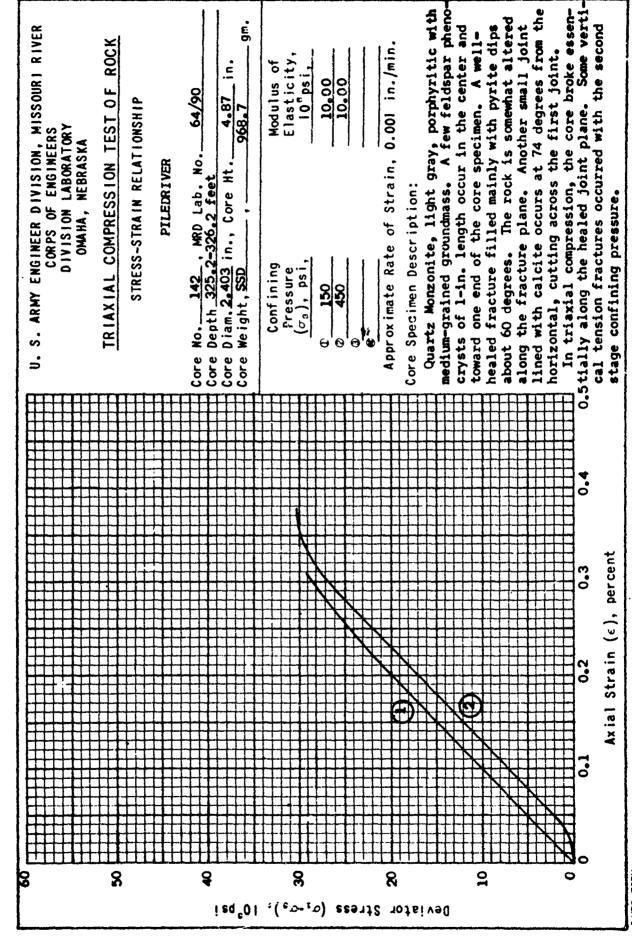
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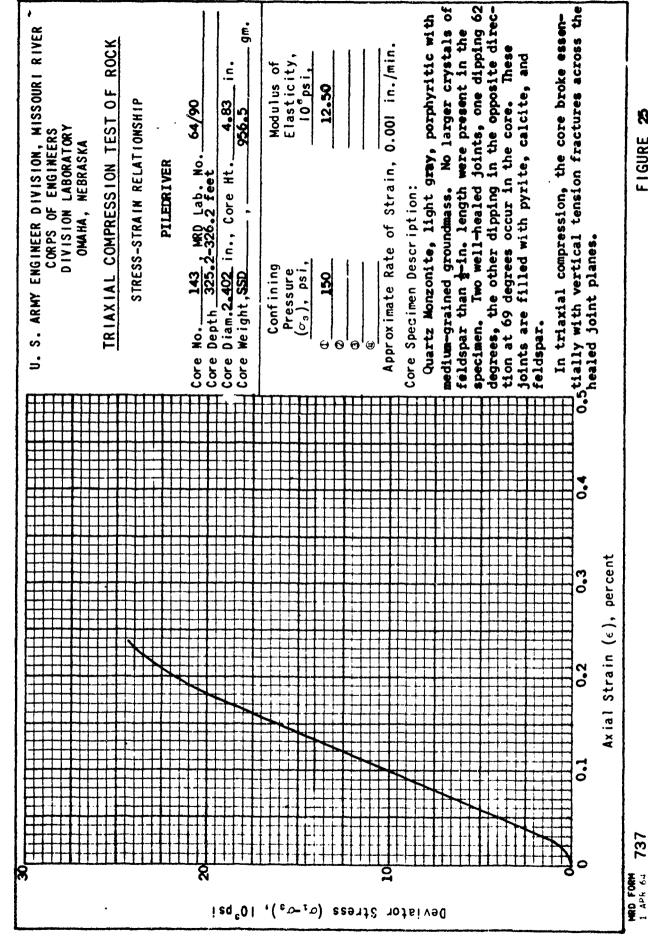


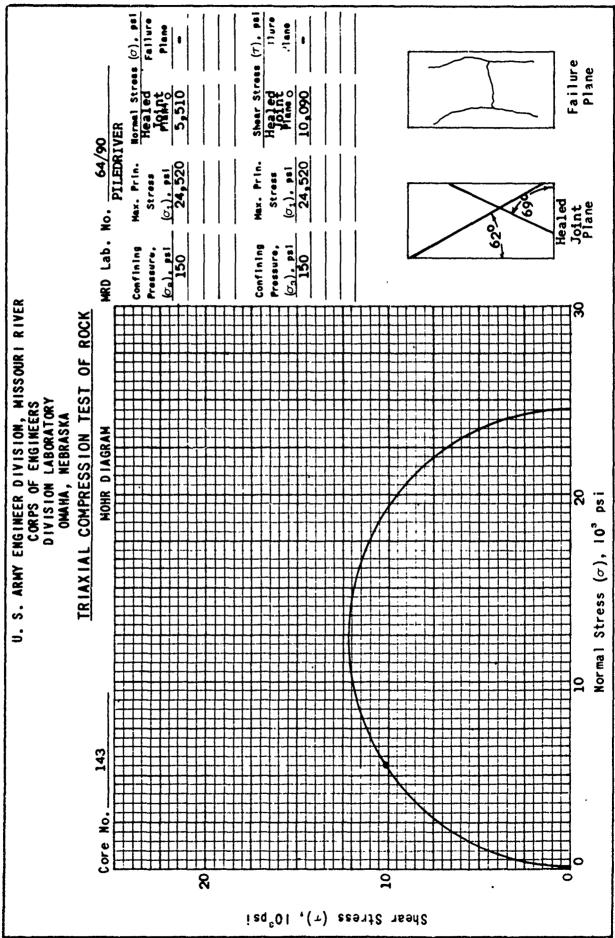


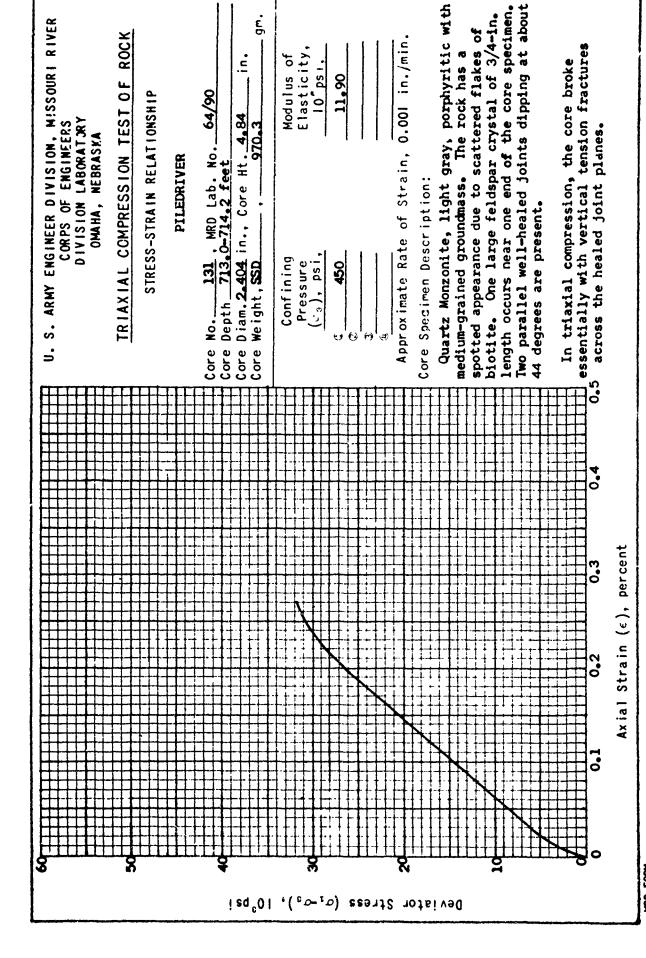
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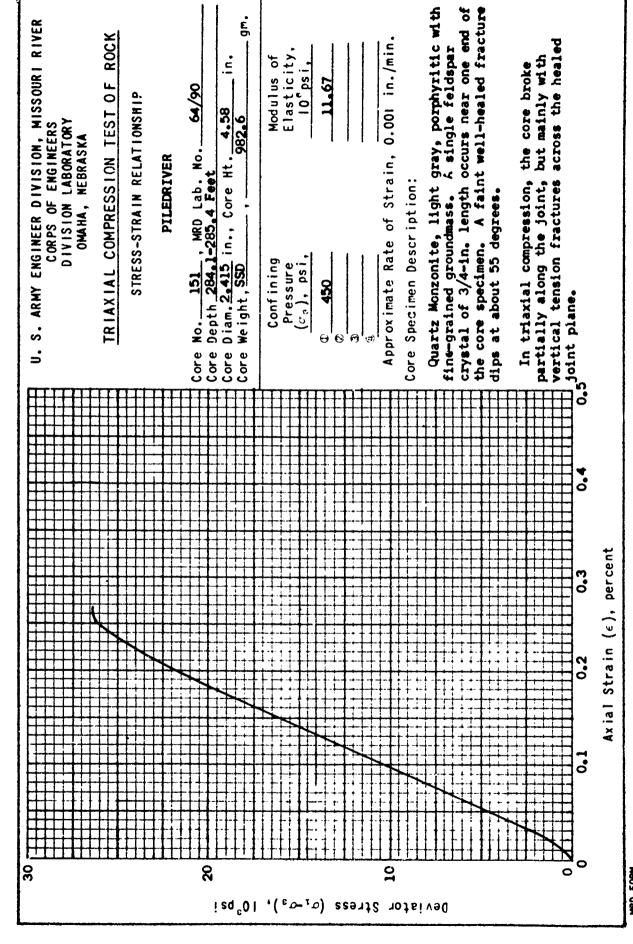
FIGURE 22



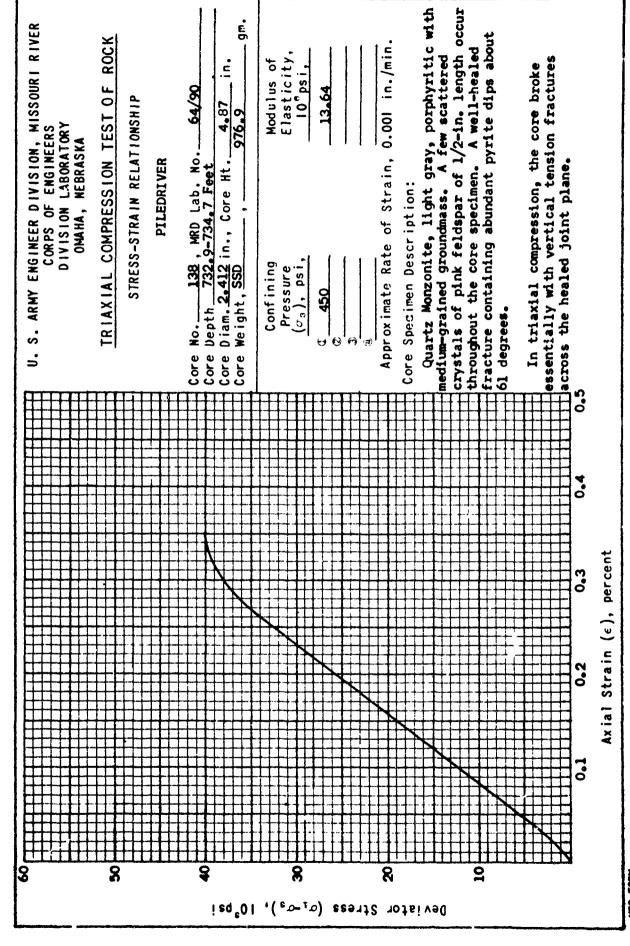




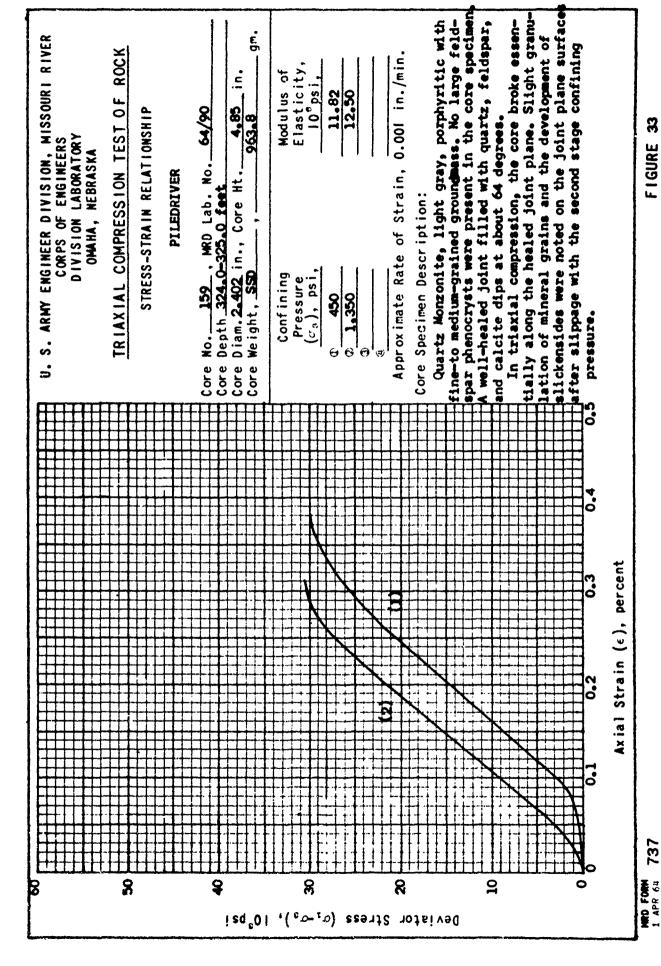




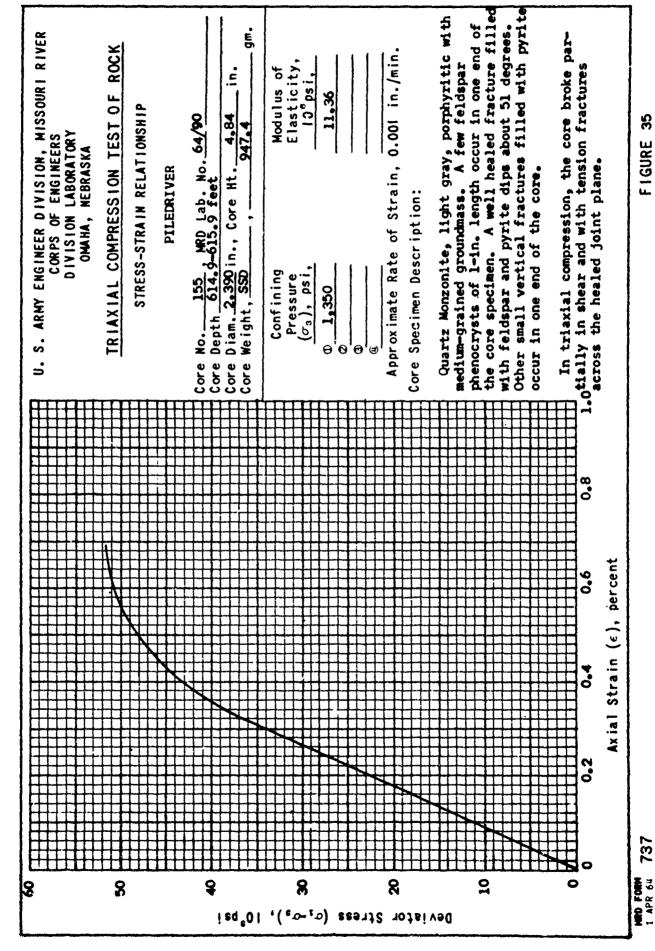
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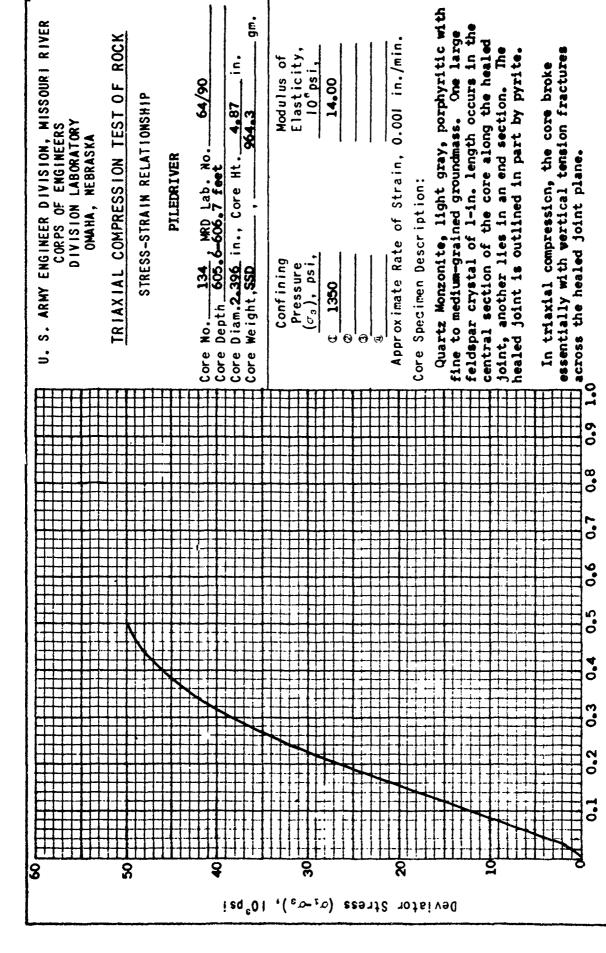
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1 APR 64 738



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37 FIGURE

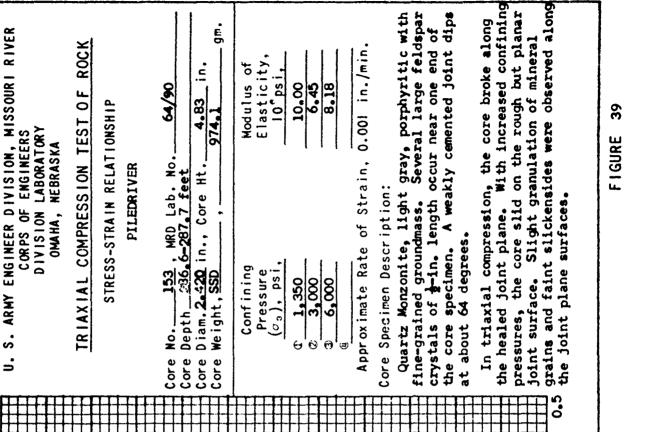
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Axial Strain (€), percent

HRD FORM 738

FIGURE 38



2

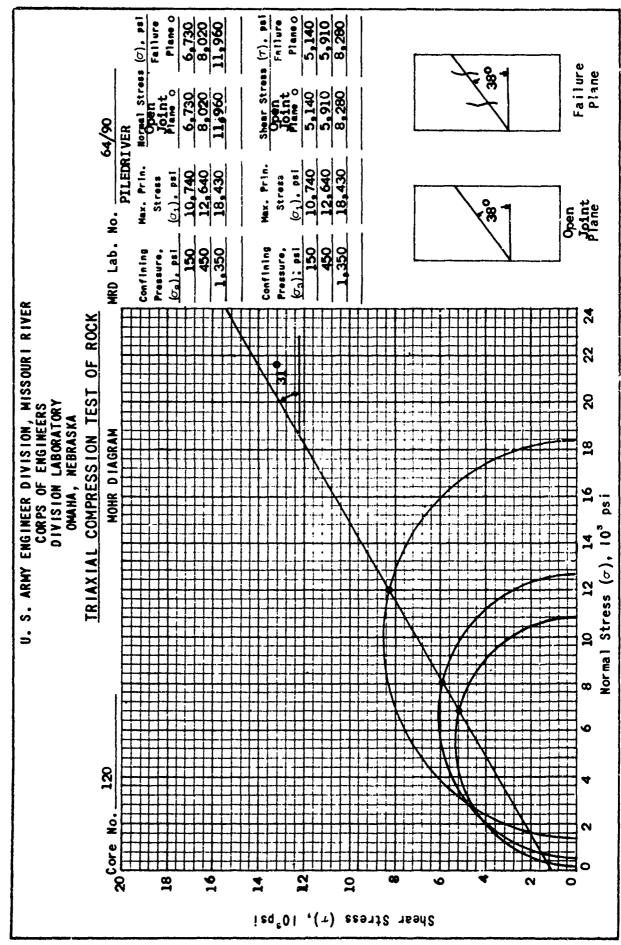
2

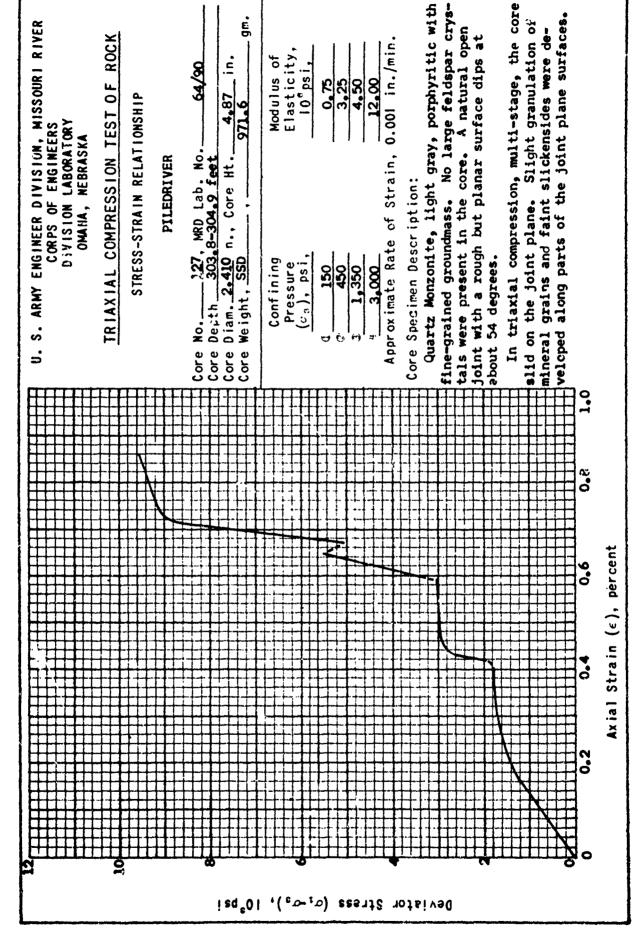
Deviator Stress $(\sigma_1 - \sigma_0)$, 10^9 psi

TAPE 64 737

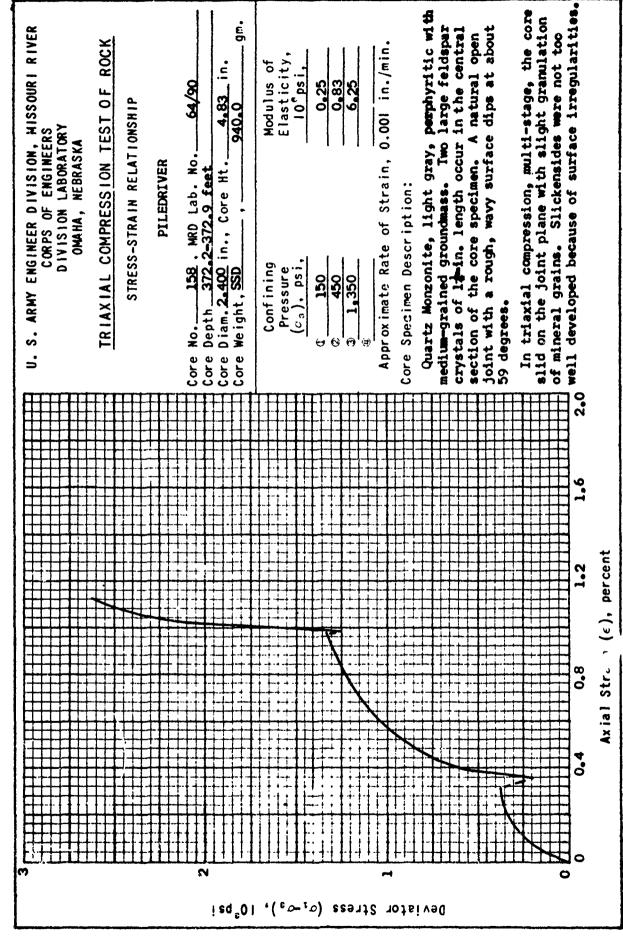
Axial Strain (<), percent

MRD FORM 738

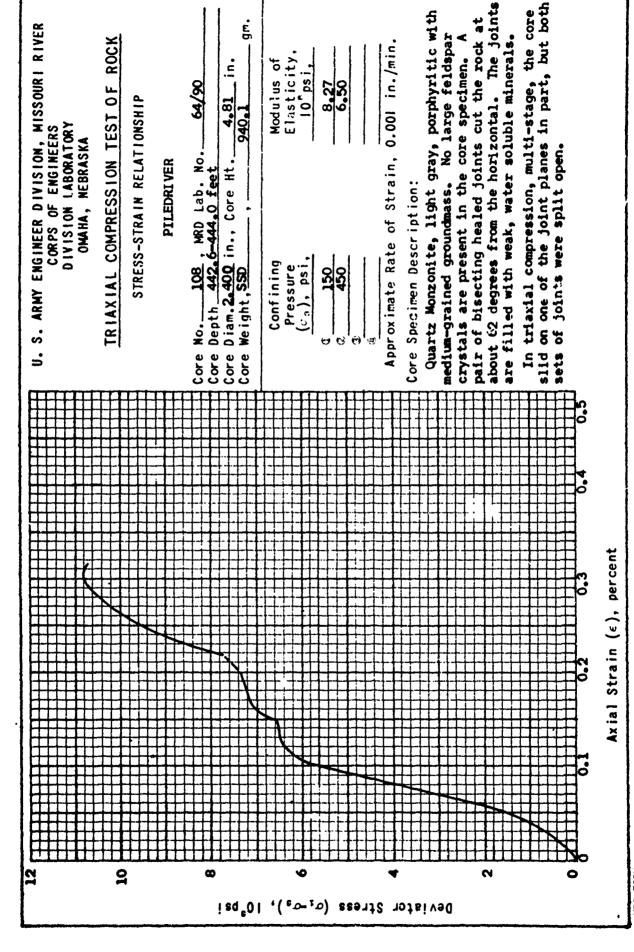


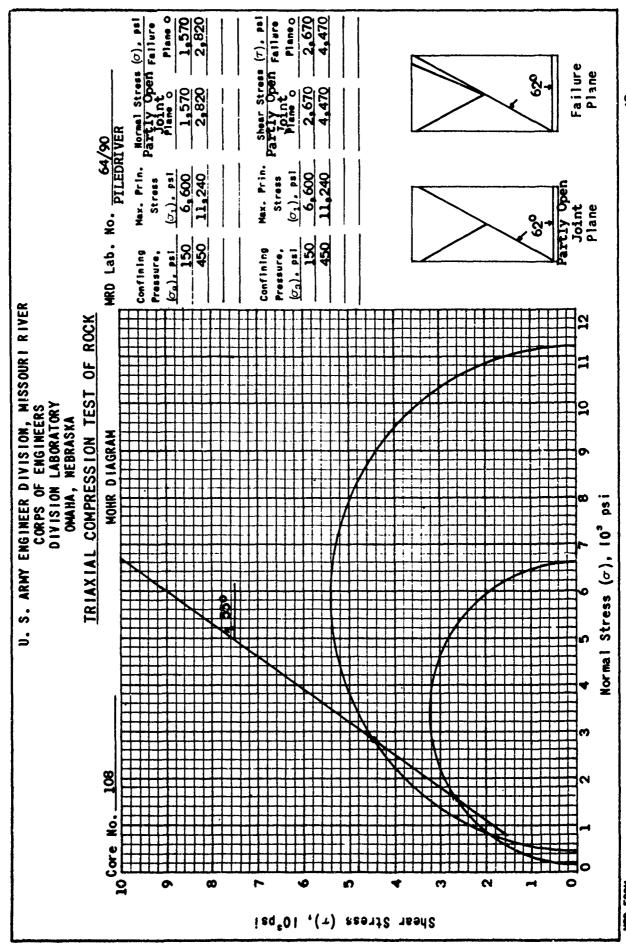


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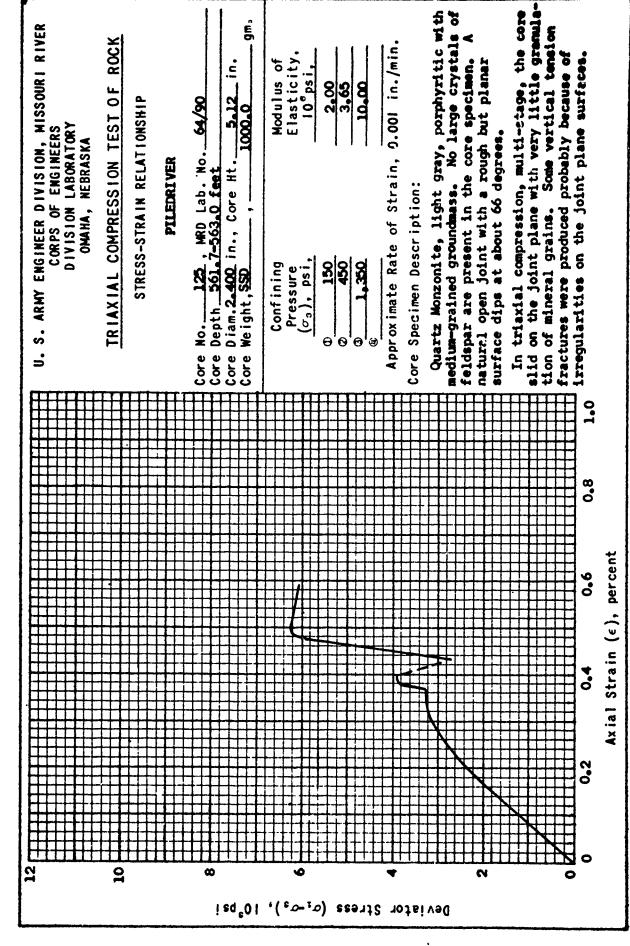
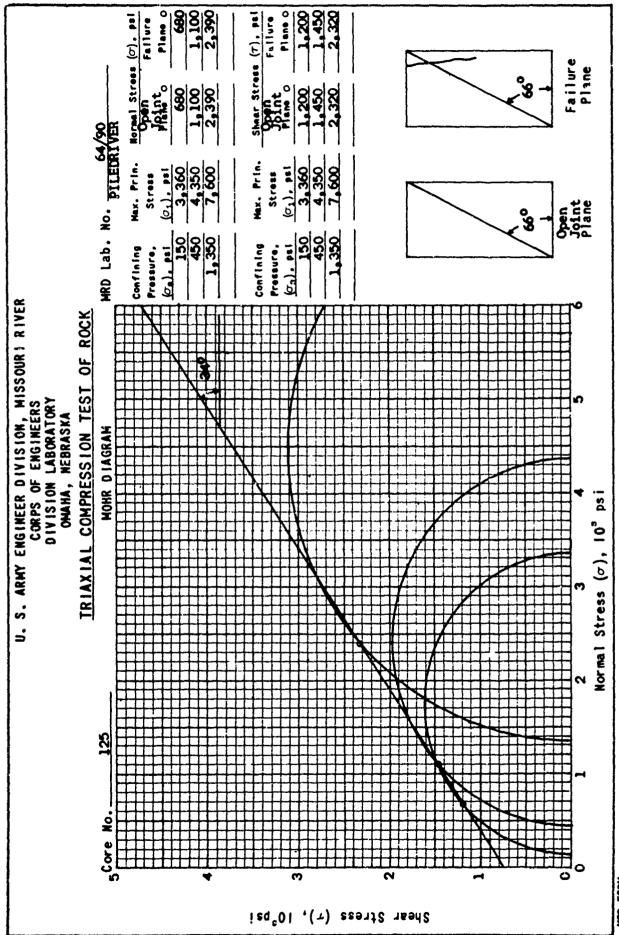
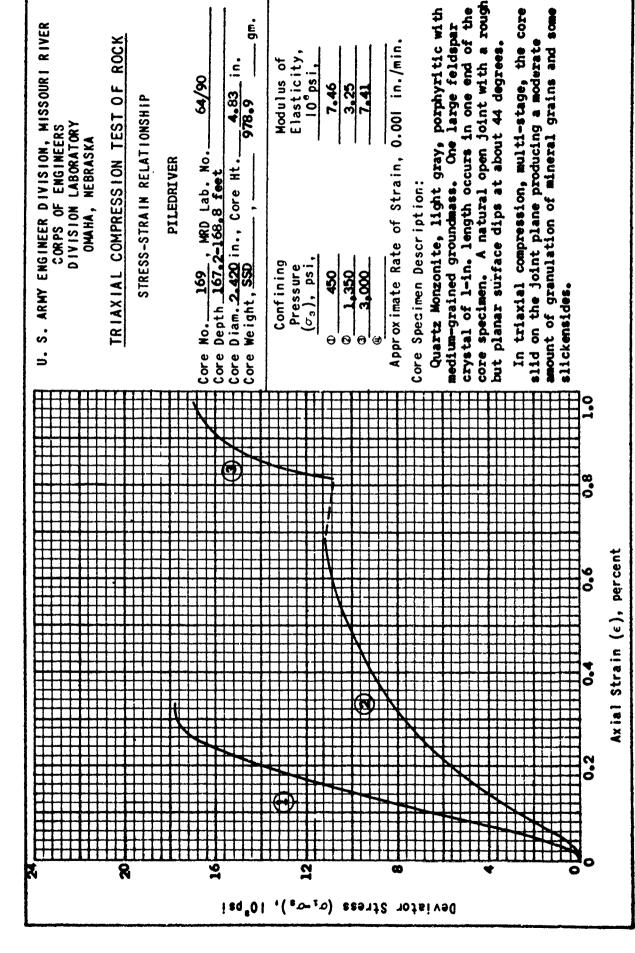
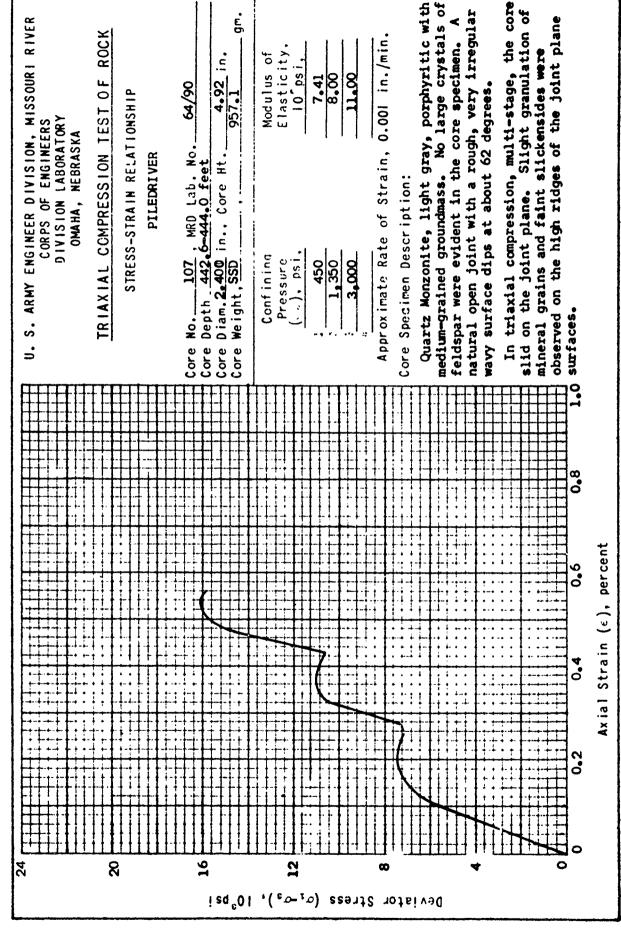


FIGURE +







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FIGURE 54

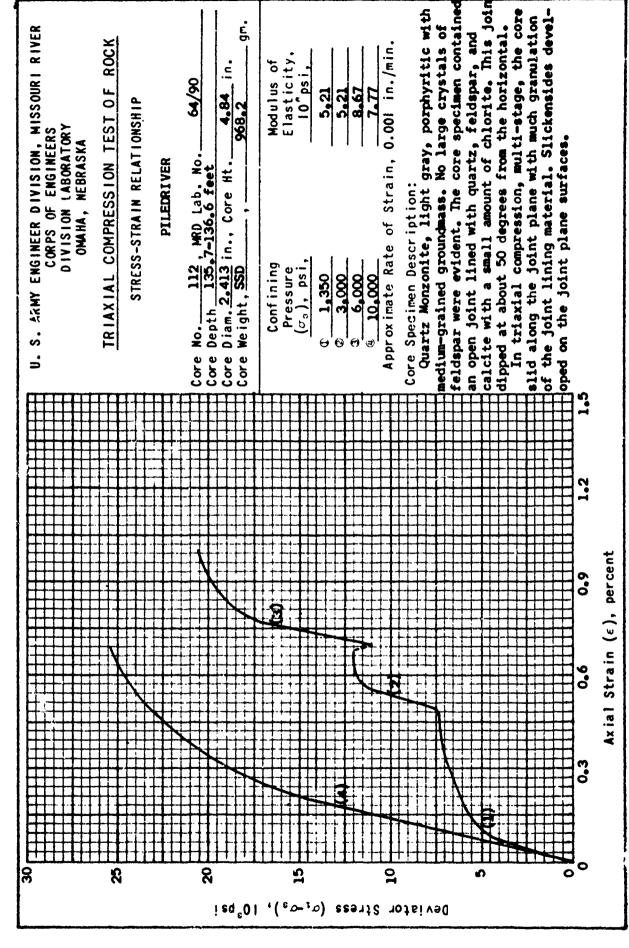
Deviator Stress $(\sigma_1 - \sigma_3)$, 10^3 psi

n

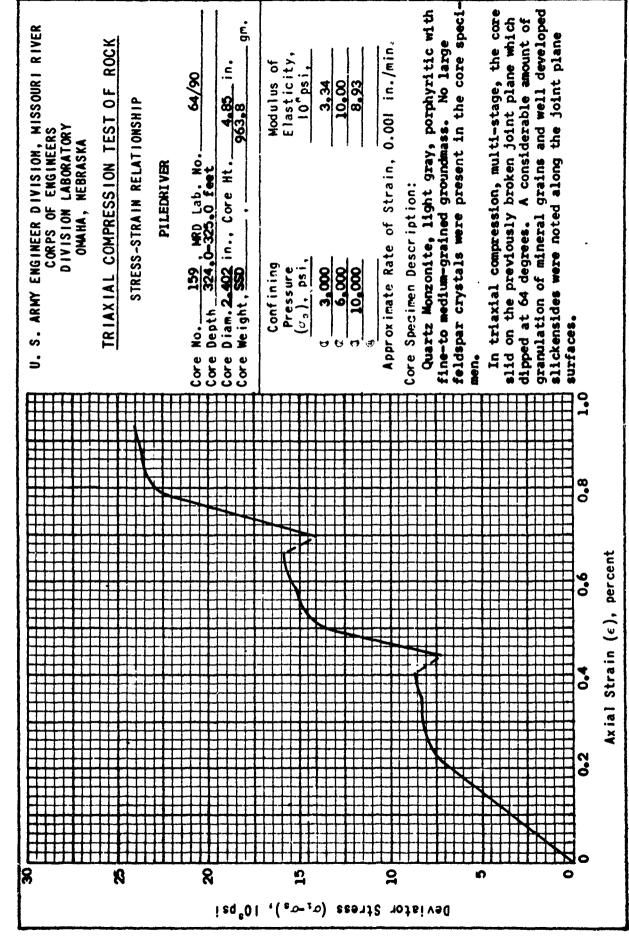
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FIGURE 55

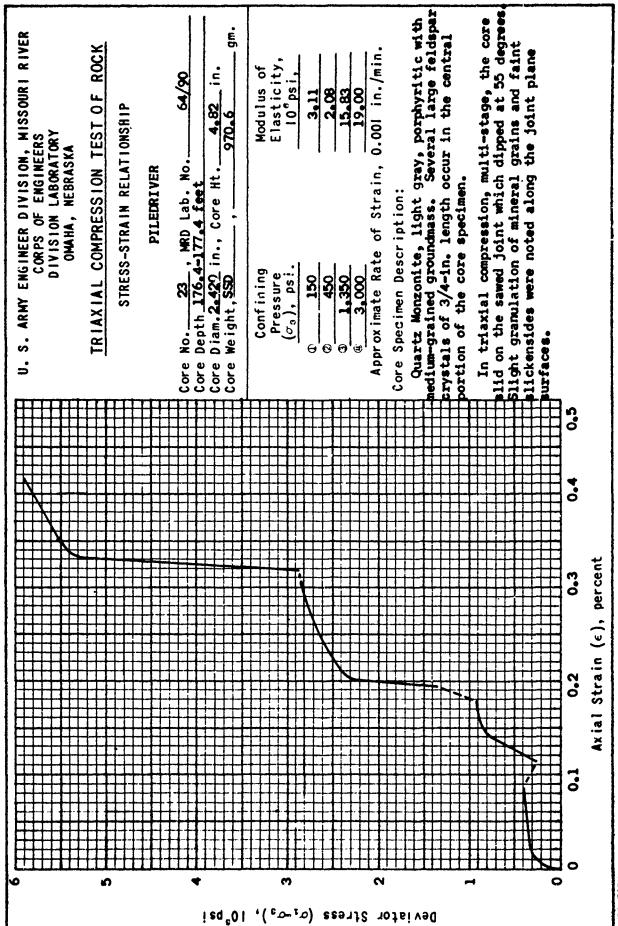
ИВО FORM 738

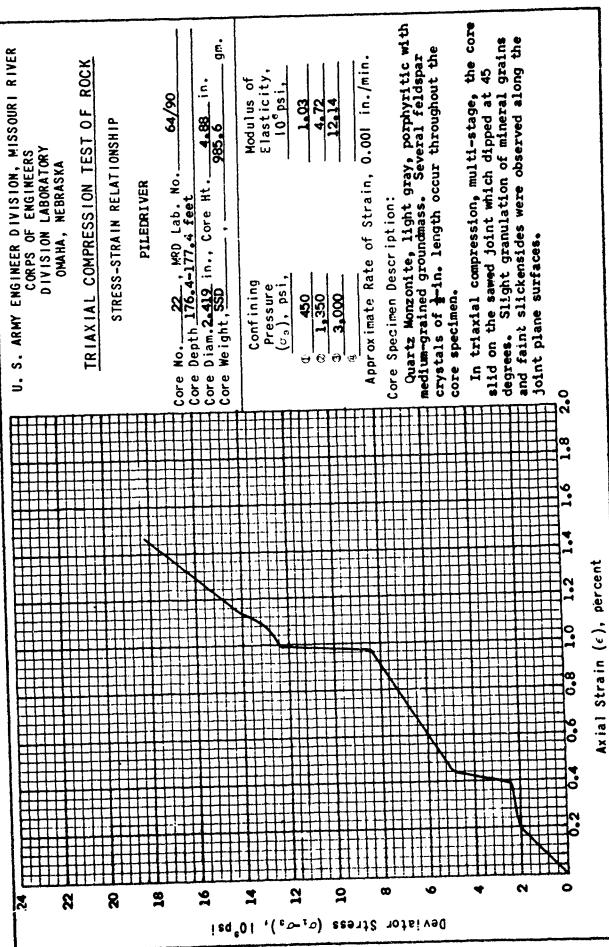


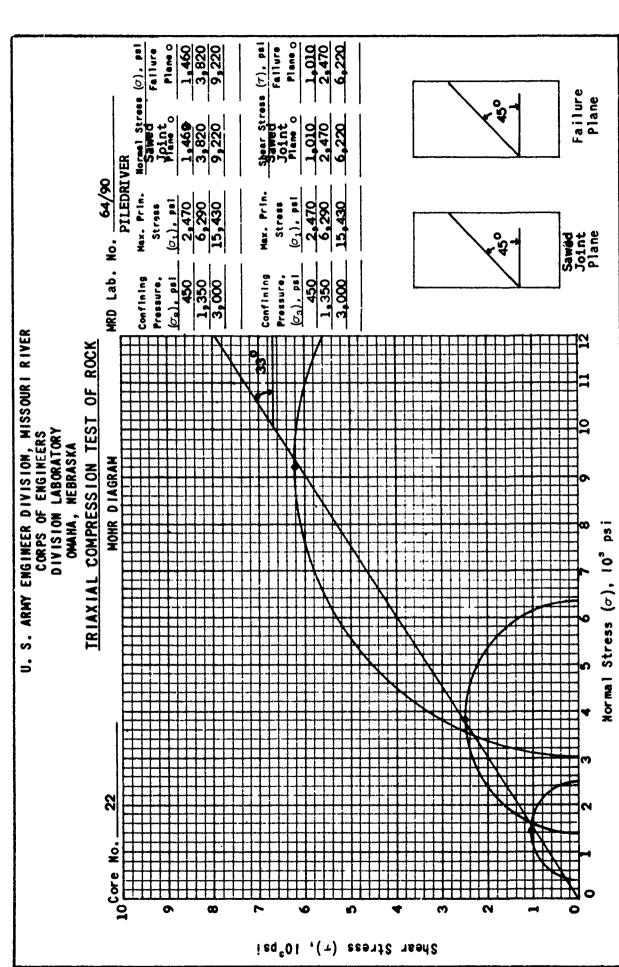
1 APR 64 738

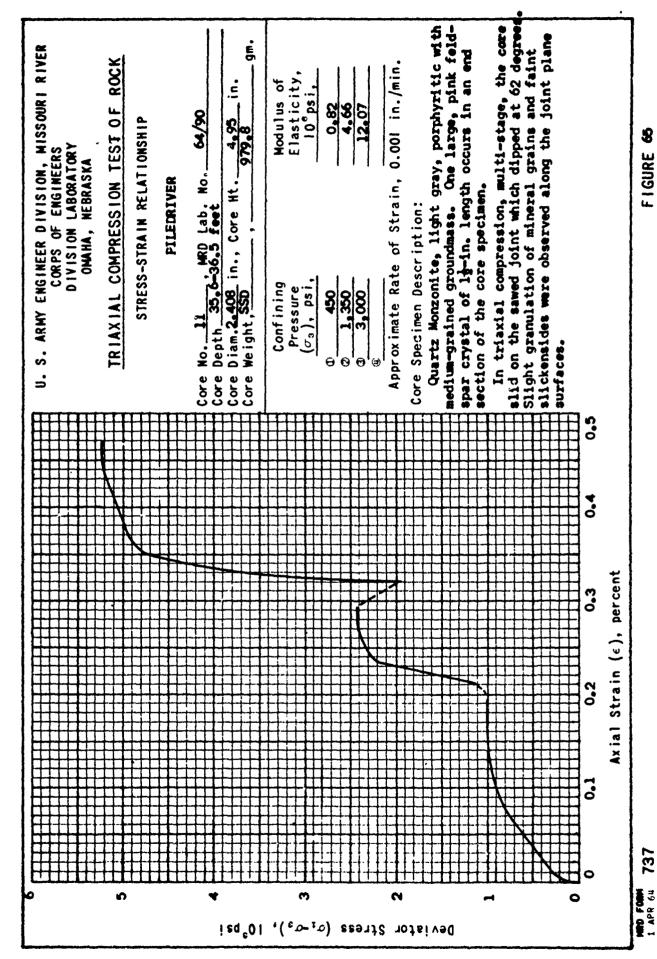


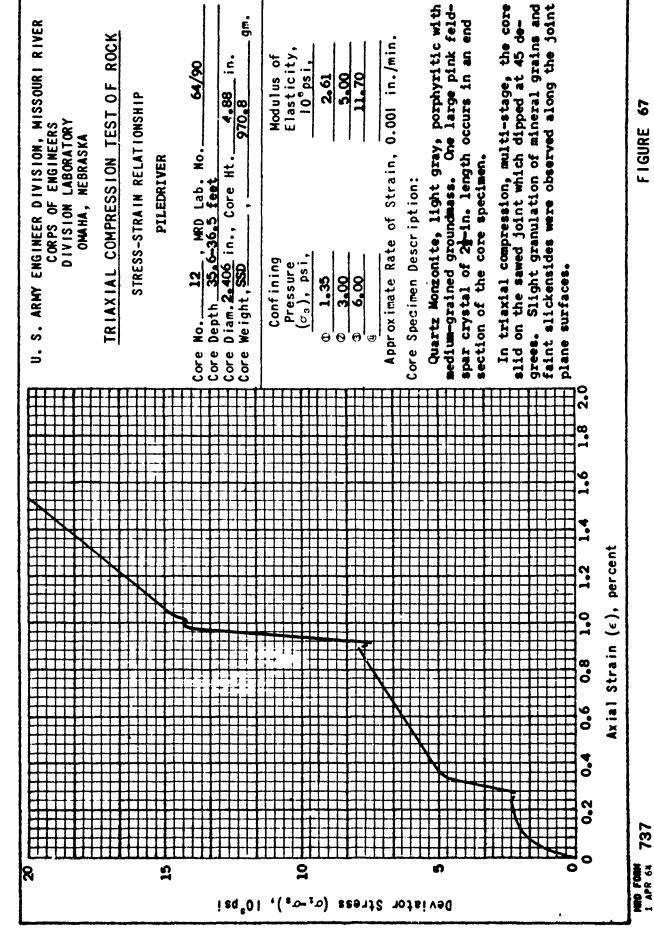
HED FORM 72R



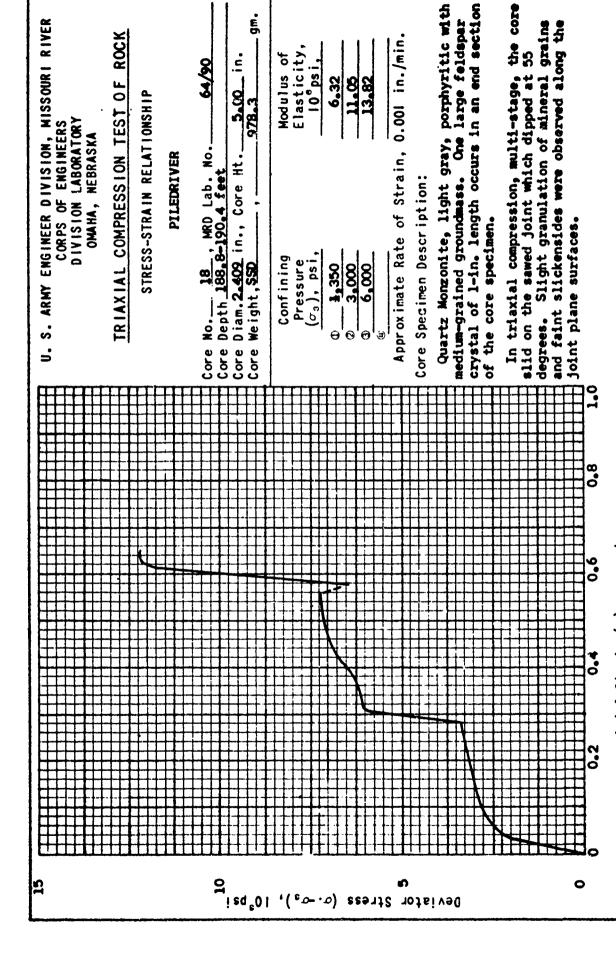








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10°psi,

11.05 6.32

13.82

FIGURE

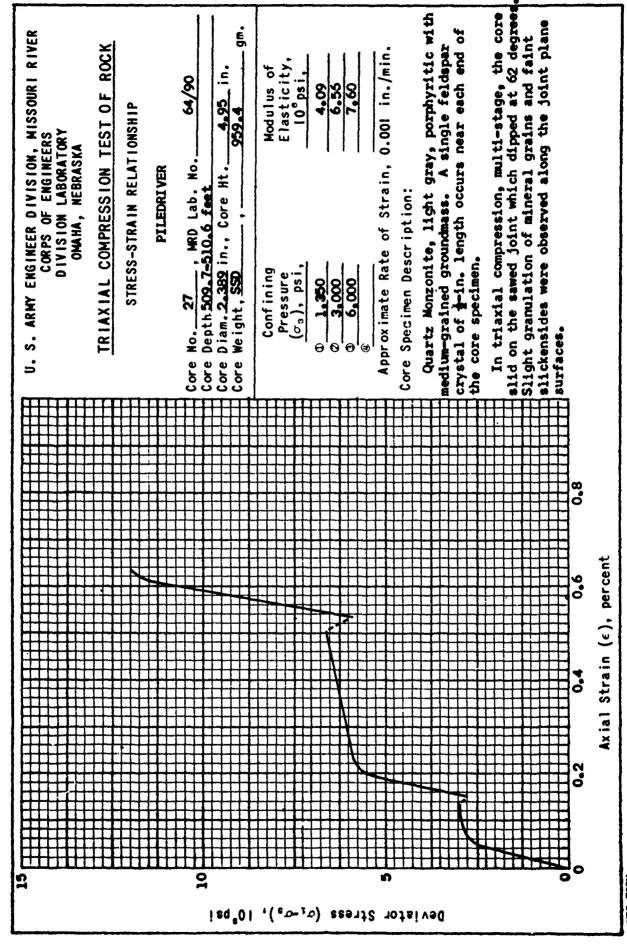
Axial Strain (ϵ) , percent

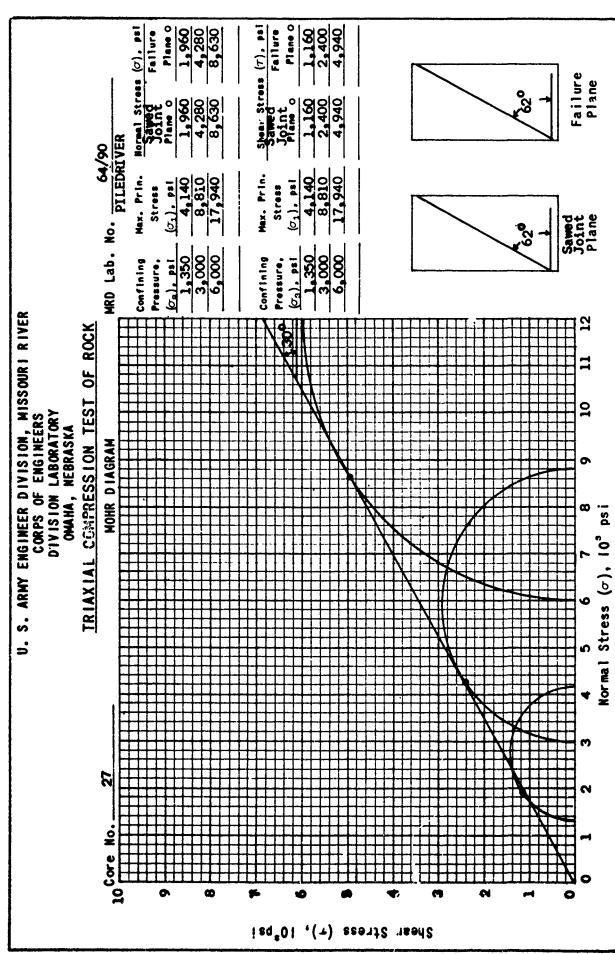
Shear Stress (7), 10³psi

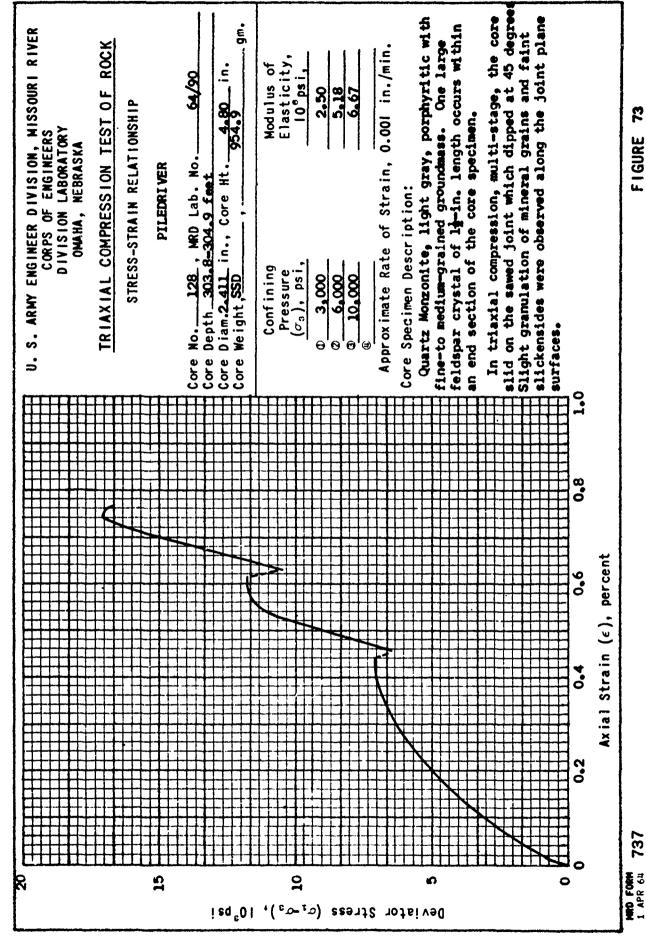
~# F

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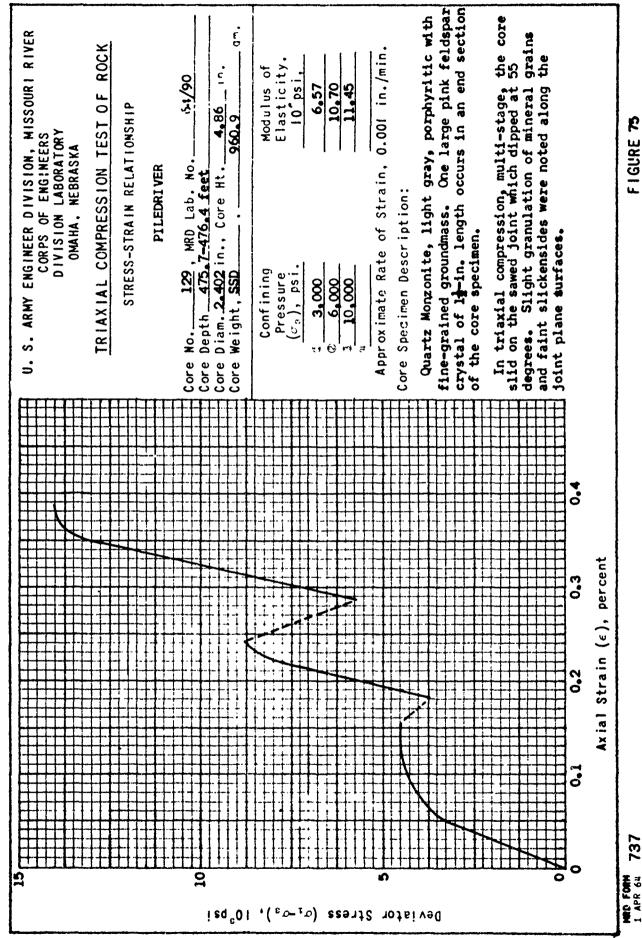
MMD FORM 738



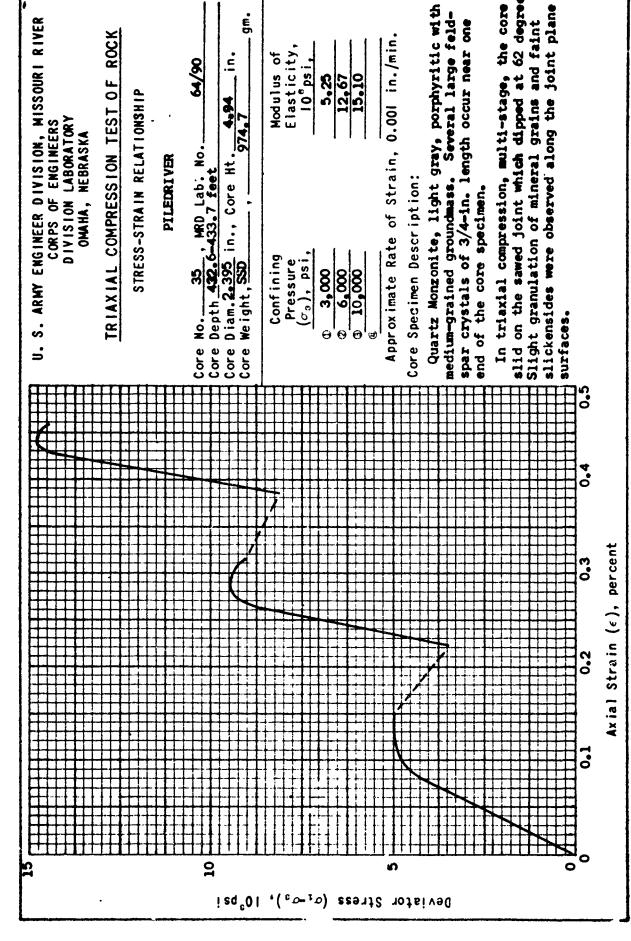




MRD FORM 738

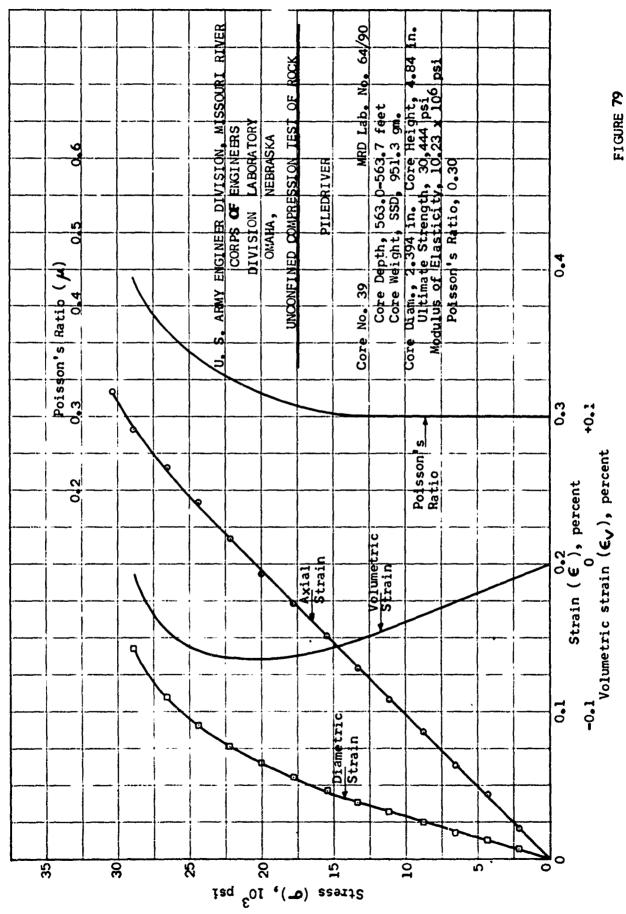


MRD FORM 738

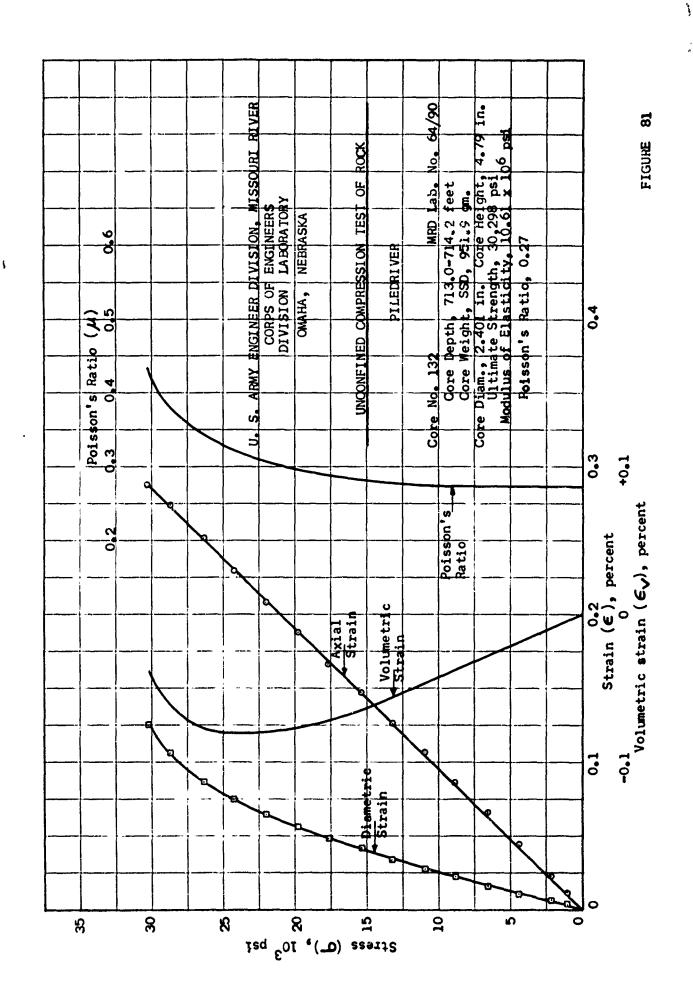


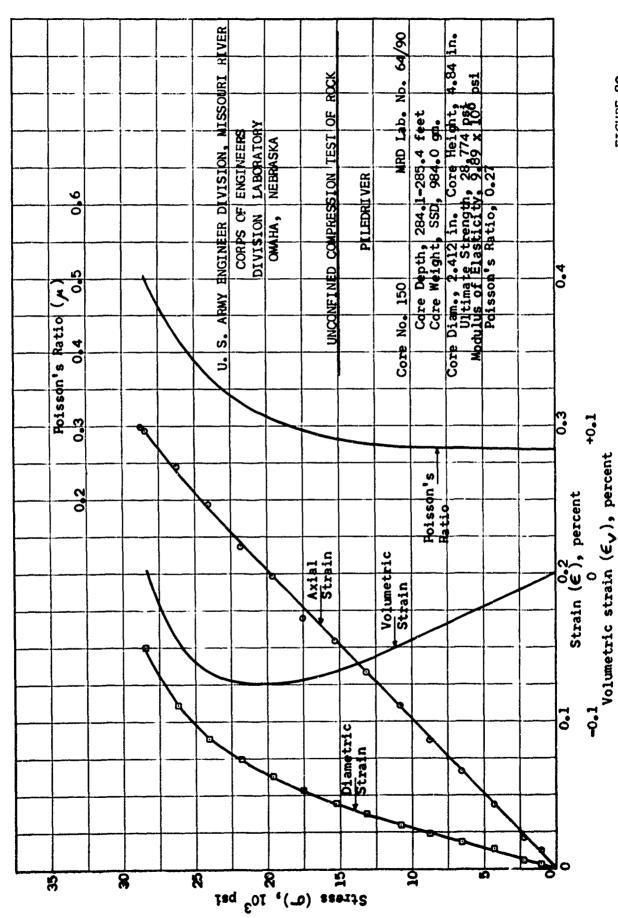
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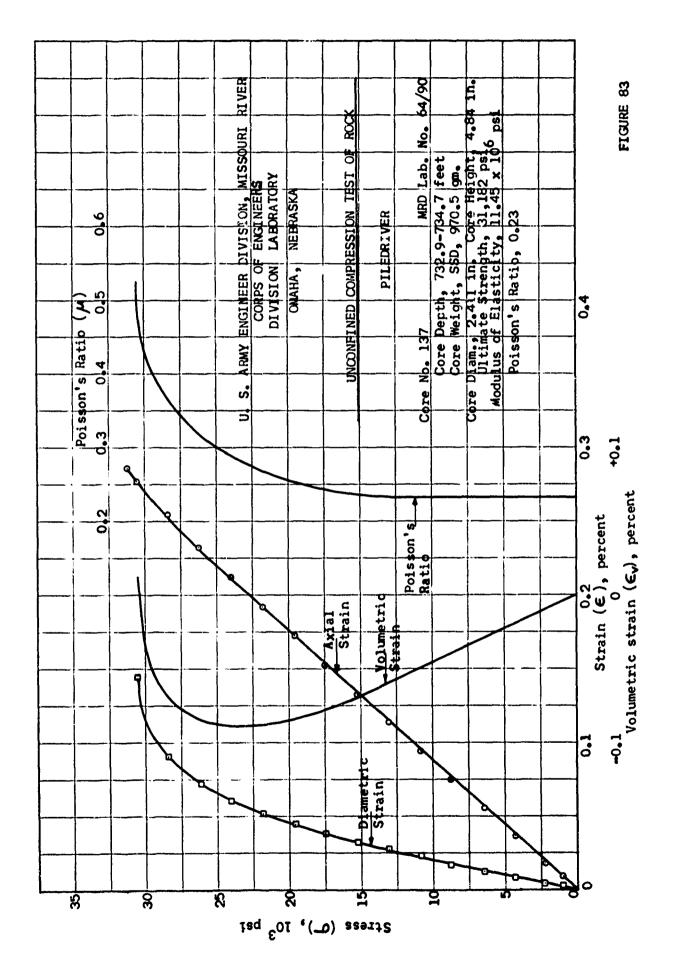
1 APR 64 738



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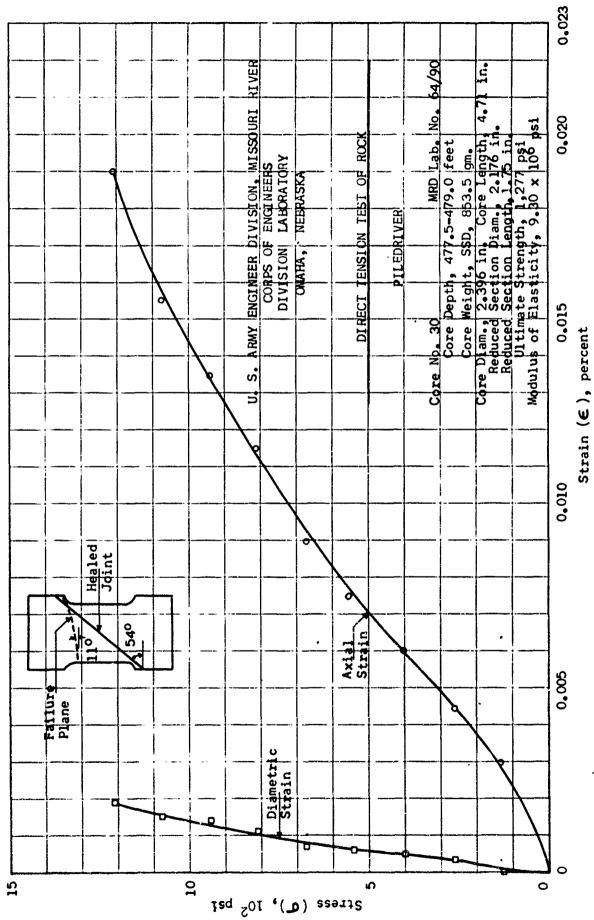
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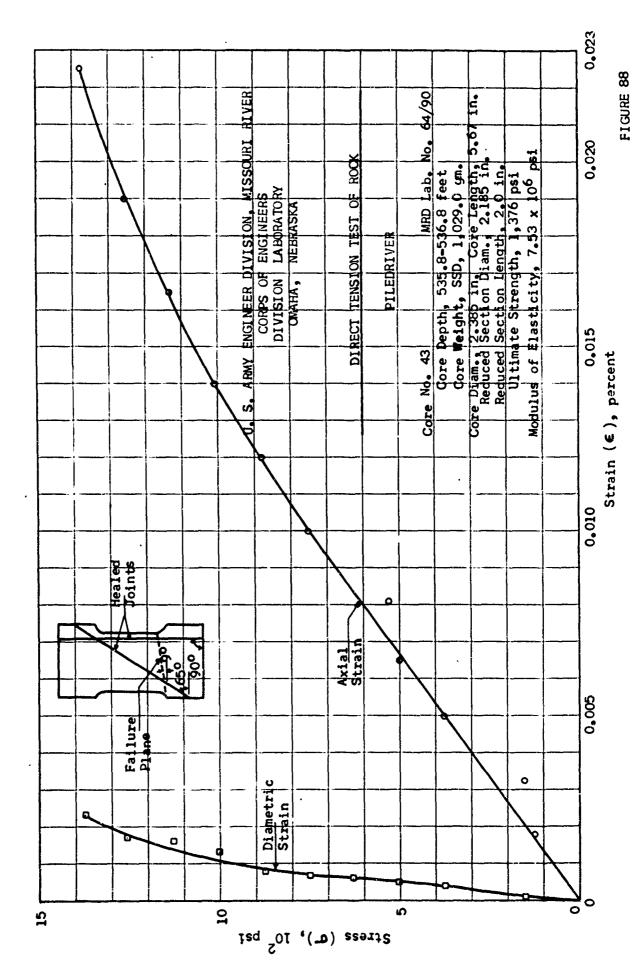
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FIGURE 86

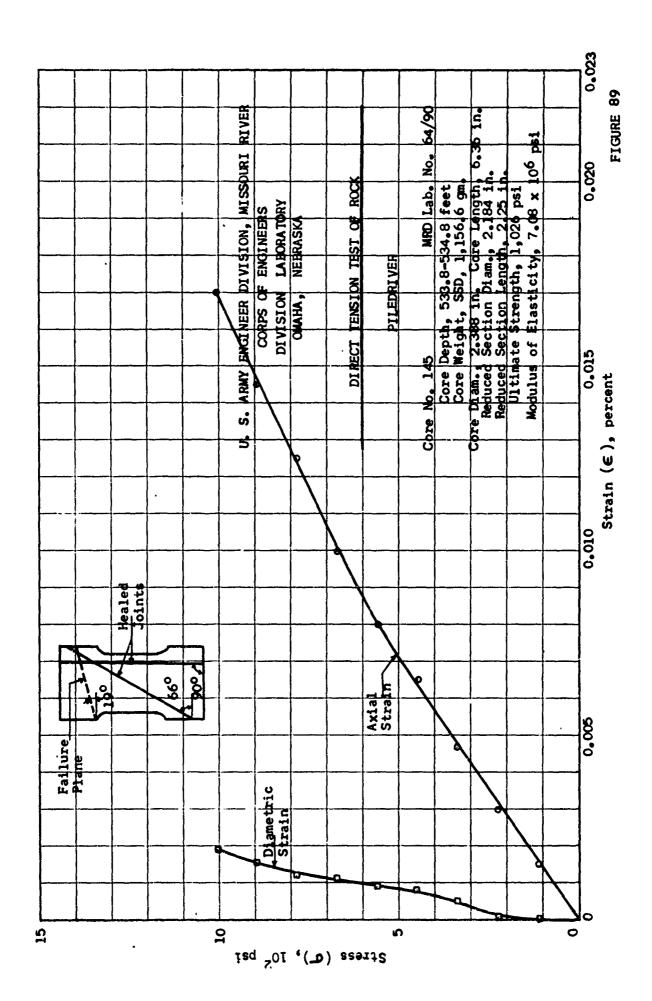


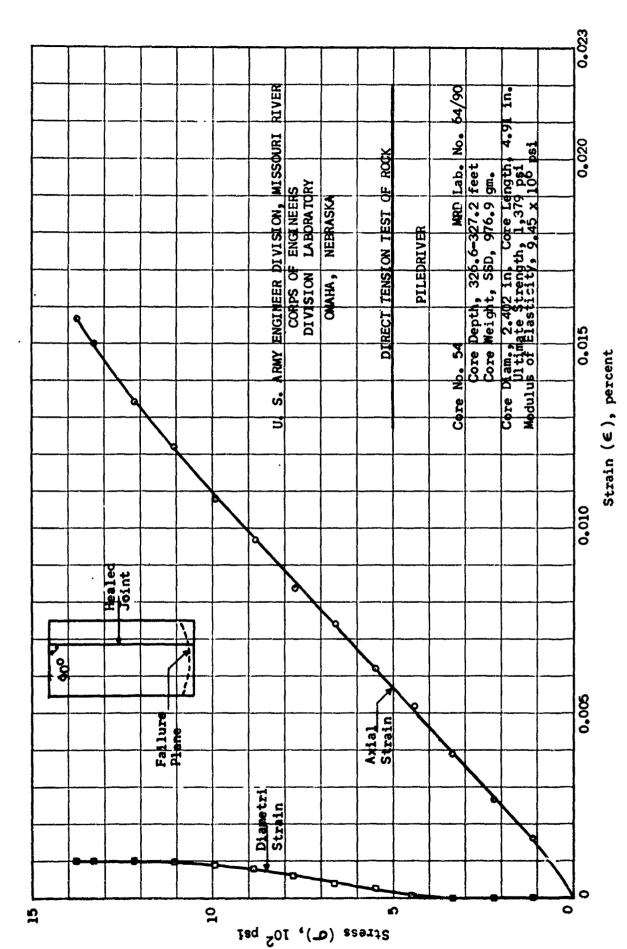


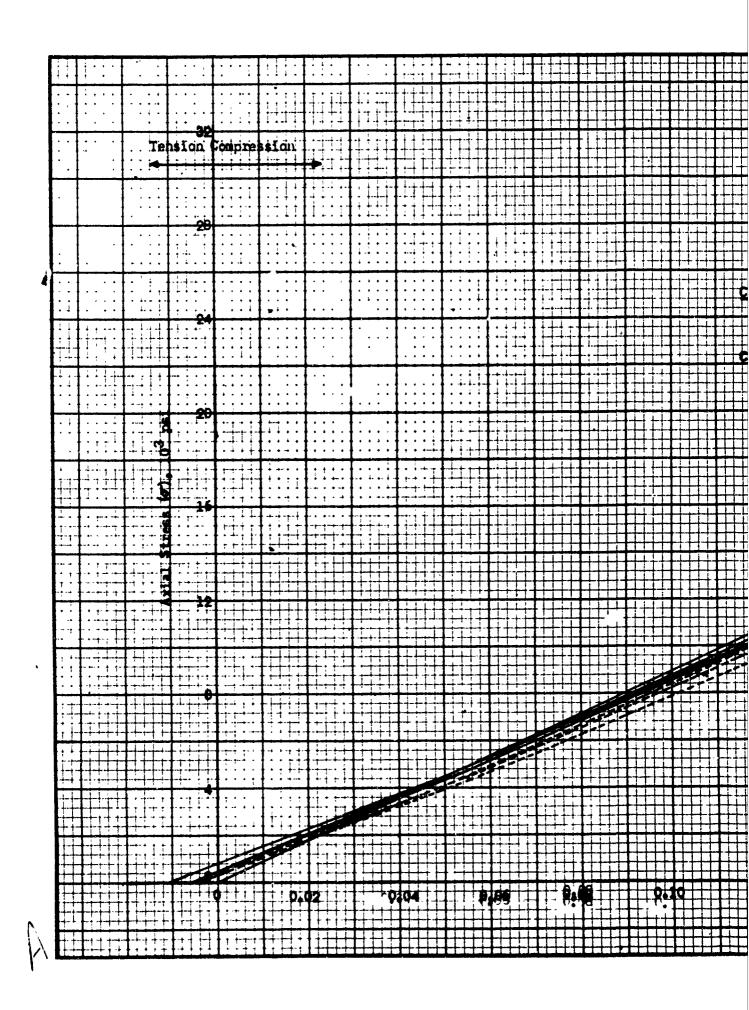
`)

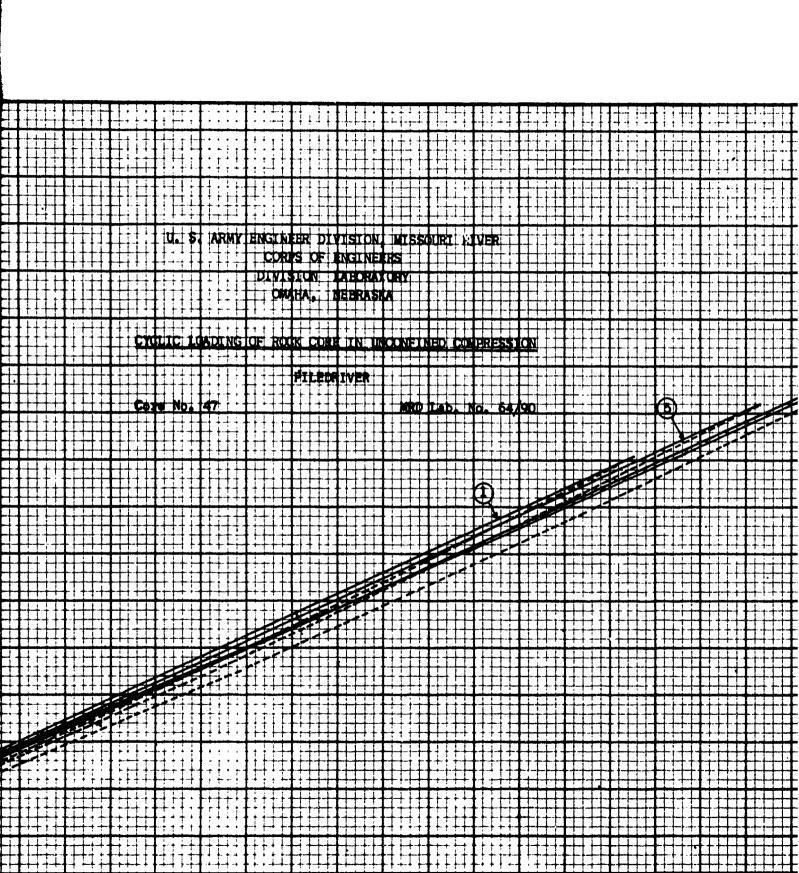
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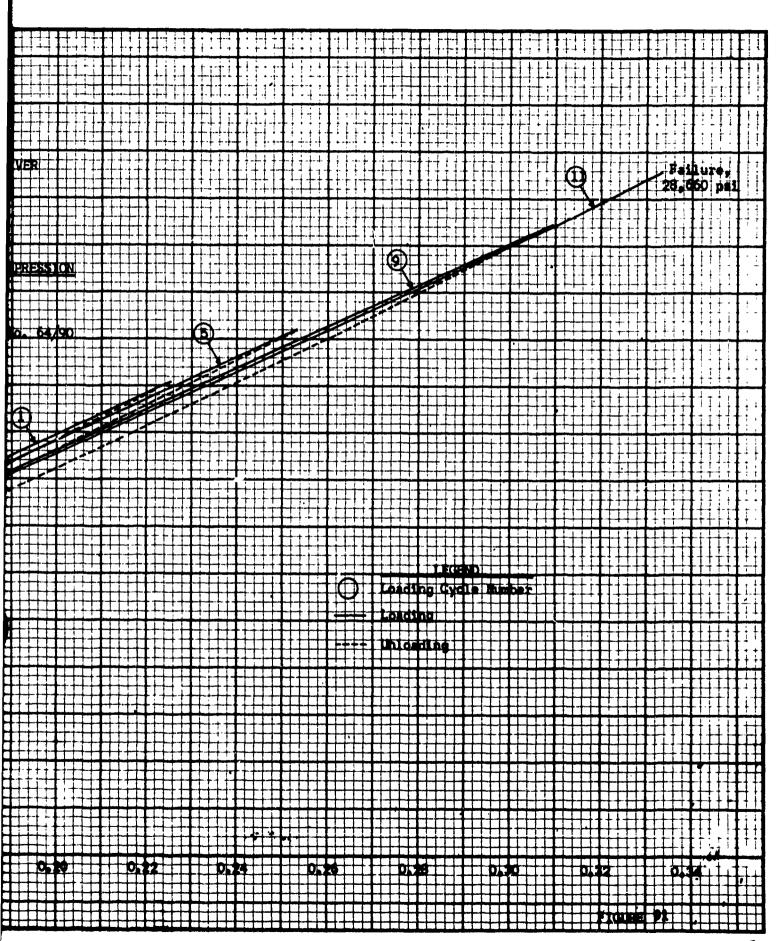
k :





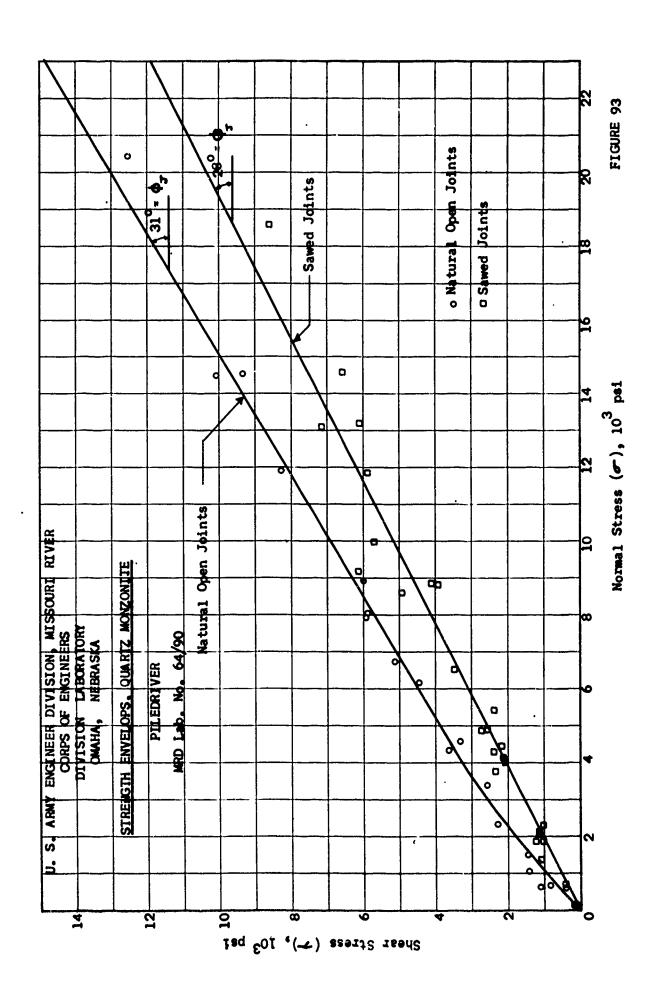






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7.

U. S. ARMY ENGINEER DIVISION, MISSOURI RIVER Sheet 1 of 2 CORPS OF ENGINEERS DIVISION LABORATORY OMAHA, NEBRASKA

TABLE 1. -- SPECIFIC GRAVITY, DENSITY, AND POROSITY OF ROCK

Piledriver			MRD Lab.	No. 64/90
		Specific	Measured Bulk	
	Core	Gravity,	Dry Density,	
	No.	Bulk SSD.	om./cu.cm.	Porosity
	سائنت			
Triaxial Compression Test	37	2.67	2.65	0.19
Specimens - Intact Rock	133	2.67	2.64	0.24
	144	2.67.	2.67	0.31
	40	2.68	2.66	0.30
	46	2.67	2.65	0.35
	69	2.70	2.68	0.21
	28	2.69	2.67	0.35
	8 3	2.67	2.67	0.19
	136	2.67	2.65	0.32
		-		• -
Triaxial Compression Test	130	2.67	2.66	0.35
Specimens - Natural Healed	105	2.67	2.67	0.22
Joints	142	2.66	2.65	0.35
	143	2.66	2.65	0.35
	131	2.67	2.66	0.35
	151	2.69	2.68	0.21
	138	2.67	2.66	0.19
	159	2.69	2.67	0.16
	155	•	•	•
	134	2.67	2.66	0.26
	153	2.70	2.67	0.20
Triaxial Compression Test	120	2.68	2.63	0.20
Specimens - Natural Open	127	2.68	2.67	0.15
Joints	158	2.66	2.63	0.33
	108	2.66	2.63	0.22
	125	2.67	2.67	0.22
	169	2.69	2.68	0.21
	107	2.66	2.63	0.22
	164	2.6 8	2.67	0.69
	112	2.69	2.64	0.24
	159	2.69	2.67	0.16
Triaxial Compression Test	23	2.69	2.65	0.22
Specimens - Sawed Joints	22	2.69	2.65	0.22
openiment outling	11	2.67	2.64	0.23
	12	2.67	2.64	0.23
	18	2.70	2.67	0.21
	27	2.68	2.65	0.25
	128	2,00	2,03	0.25
	129	2.67	2.65	0.22
	35	2.68	2.65	0.20
	33	2000	2,00	V

	Core	Specific Gravity, Bulk SSD.		Porosity
Unconfined Compression Test Specimens - Intact Rock	39 86 132	2.67 2.67 2.68	2.67 2.65 2.64	0.22 0.23 0.40
Unconfined Compression Test Specimens - Natural Healed Joints	150 137	2.67	2.68 -	0.21
Direct Tension Test Specimens - Intact Rock	9 88 126	2.68 2.67	2.65 2.67	0.29 0.22
Direct Tension Test Specimens - Natural Healed Joints	30 43 145 54	2.68 2.67 2.68 2.68	2.67 2.65 2.64 2.66	0.36 0.18 0.31 0.20
Cyclic Loading Unconfined Compression Test Specimen - Intact Rock	47	2.67	2.55	0.35

U. S. ARMY ENGINEER DIVISION, MISSOURI RIVER CORPS OF ENGINEERS DIVISION LABORATORY OMAHA, NEBRASKA

TABLE 2. -- CYCLIC LOADING OF HOCK CORE IN UNCONFINED COMPRESSION

PILEDRIVER

Core No. 47

MRD Lab. No. 64/90

		Ϋ́ V	Axial Strain for Indicated Loading Cycle.	in for I	ndicated	Loading		micro-inches/inch	ches/inc	adi		
Stress, ksi	4	2	6	4	2	ها	7	8	٩	9	1	ا
c	c	+ 45	\$	+ 45	+	+ 55	+	\$	+ 62	+ 74	+	118
021	- 135	202	• 82 82	- 155	82	83	- 81	2	- 79	- 5	ŧ	45
0.00		- 228	280	- 227	. 220	- 225	- 230	- 212	- 230	- 230	ı	Š
4.478	200	505	- 442	510	200	206	- 512	- 502	- 515	- 520	•	497
6.717	- 765	- 785	. 757	- 766	- 770	- 770	- 781	- 770	- 795	- 798		275
057	1007	1030	0101-	-1020	-1022	-1028	-1035	-1030	-1060	-1065		1040
11 196	-1247	-1279	-1263	1200	-1272	-1280	-1290	-1290	-1315	-1320		1310
13.425	200	+1545	-1520	-1550	-1535	-1545	-1554	-1558	-1585	-1590	ŧ	1580
15.674	-1750	-1780	-1770	-1793	-1780	-1795	-1800	-1815	-1835	-1850	•	1839
17.913	-2000	-2020	-2016	-2040	-2028	-2045	-2055	-2060	-2085	-2100		2095
20.152	-2255	-2280	-2260	-2285	-2280	-2285	-2300	-2306	-2355	-2350	ŧ	2342
701 201) } !	-2535	-2531	-2540	-2555	-2560	-2590	-2600	1	2605
24 621)		-2795	-2805	-2809	-2840	-2860	•	2855
25.870							l	-3070	-3095	-3107		3100
20.109										-3355	ı	3336
28,660											Fai	lure
26.870										-3120		
24.631								-2845	-2870	-2895		
22,391						-2565	-2555	-2618	-2640	-2655		
20,152				-2305	-2295	-2335	-2340	-2376	-2400	-2410		
17.913	-2030	-2040	-2035	-2075	-2073	-2105	-2110	-2139	-2165	-2166		
15,674	-1795	-1820	-1805	-1840	-1837	-1865	-1870	-1895	-1916	-1920		
13,435	-1555	-1580	-1560	-1595	-1591	-1615	-1626	-1646	-1665	-1664		
11.196	-1310	-1332	-1315	-1342	-1343	-1360	-1365	-1380	-1400	-1400		
8,957	-1063	-1080	-1060	-1090	-1086	-1095	-1100	-1096	-1128	-1115		
6,717	- 805	- 826	- 805	- 830	- 815	- 833	838	- 845	- 828	- 845		
4.478	- 540	- 559	- 535	- 560	- 545	- 555	- 555	- 560	- 570	- 556		
2,239	- 257	- 265	- 245	- 267	- 250	- 260	- 250	- 250	, 88	- 236		

gage	ate strain	es indicate	positive values		compression,	gage com	strain	indicate	values	Negative tension.	Notes
	<u>s</u>	ծ •	£ +	÷	÷ 26	\$	+	8	+ 35	+ 5	0
	2	8 1	ନ 	• 32	101 -	- 95	- 115	- 95	- 115	- 110	1.120
	- 236	8	520	- 520	- 260	- 250	- 267	- 245	- 265	- 257	2.239
	- 556	- 570	. 26	- 555	- 555	- 545	- 560	- 535	- 559	- 540	4.478
	- 845	828	- 845	838 •	- 833	- 815	- 830	= 805	- 826	- 805	6,717
	-1115	-1128	-1096	-1100	-1095	-1086	-1090	-1060	-1080	-1063	8,957
	-1400	-1400	-1380	-1365	-1360	-1343	-1342	-1315	-1332	-1310	11,196
	-1664	-1665	-1646	-1626	-1615	-1591	- 1595	-1560	-1580	-1555	13,435
	-1920	-1916	-1895	-1870	-1865	-1837	-1840	-1805	-1820	-1795	15,674
	-2166	-2165	- 2139	-2110	-2105	-2073	-2075	-2035	-2040	-2030	17,913
	-2410	-2400	-2376	-2340	-2335	-2295	-2305				20,152
	-2655	-2640	-2618	-2555	-2565						22,391
	-2895	-2870	-2845								24.631
	-3120										26.870

See Figure 91 for stress-strain curve.

U. S. ARMY ENGINEER DIVISION, MISSOURI RIVER
CORPS OF ENGINEERS
DIVISION LABORATORY
OMAHA, NEBRASKA

TABLE 3. - CYCLIC LOADING OF ROCK CORE IN UNCONFINED COMPRESSION

Core No. 47			PIL	PILEDRIVER	æ				MRD	Lab. N	WRD Lab. No. 64/90
	70	metri	Stra	in for	Indic	ited Lo	paipe	Cycle	micro-1	Diametric Strain for Indicated Loading Cycle, micro-inches/inch	ache
Ksie	4	4	메	4	서	۵	7	ᆔ	어	9	4
0	0	8	78	8	100	120	4	163	222	251	522
1.120	32	66	121	127	137	154	179	197	252	58	556
2.239	8	127	150	148	169	187	214	33	231	326	595
4.478	121	161	219	217	237	257	588	8	370	4 08	689
6,71,7	180	260	286	287	310	328	362	385	452	497	796
8,957	242	327	321	357	377	403	439	461	540	290	914
11,196	310	363	450	432	451	474	517	543	6 29	689	1046
13,435	386	468	492	506	528	554	6	630	725	796	1194
15.674	465	539	565	578	3	629	8 9	111	8 8 8	868	1343
17,913	553	910	6 33	656	<i>6</i> 77	705	765	794	913	1004	1496
20,152	652	06 9	715	728	754	784	849	880	101	1107	1646
22,391				817	845	864	934	696	503	1219	1819
24.631						6	1030	1060	1218	1340	2002
26,870								1242	1390	1510	2244
29.109										2029	2946
28,660											Failure
26,870										2130	
24.631								1230	1365	2112	
22,391						942	686	1169	1300	2020	
20,152				116	792	688	931	1099	1226	1917	
17,913	612	2	670	717	733	827	867	1025	1145	1806	
15,674	561	8 8	611	939	929	765	8	952	1001	1687	
13,435	503	230	551	594	611	694	724	8	7.16	1554	
11.1%	439	466	88	522	541	919	6 42	775	872	1405	
8,957	374	3 3 3	419	453	466	236	560	579	766	1244	
6,717	808	325	347	377	391	454	473	576	653	1080	
4,478	237	220	271	297	308	365	381	468	531	6	
2,239	8	170	190	215	223	270	283	351	4	718	
1.120	121	132	120	171	179	223	236	8	8	627	
•	81	93	110	13 5	139	175	188	238	280	238	

Note: All strain values are positive, indicating strain gage tension.

See Figure 92 for stress-strain curve.